

# Mechanistic, interfacial and modeling studies of electro-precipitation during electrochemical advanced oxidation process in a thin-film reactor: case study with reclaimed municipal wastewater

Faidzul Hakim Adnan

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# **Thèse**

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## DOCTEUR DE L'UNIVERSITÉ DE LORRAINE

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par Mohd Faidzul Hakim MOHD ADNAN

Mechanistic, interfacial and modeling studies of electroprecipitation during electrochemical advanced oxidation process in a thin-film reactor: case study with reclaimed municipal wastewater

#### le 16 décembre 2021

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-Faidzul-

#### **ABSTRACT**

Electrochemical advanced oxidation processes (EAOPs) have been receiving recognition to efficiently eliminate organics, especially micropollutants, in wastewater owing to continuous production of potent, quasi non-selective hydroxyl radicals in situ. The more recent application of thin-film reactor within the framework of EAOPs has spurred the capability to treat wastewater whereby mass transport and redox reactions are intensified. It comes with huge cut in total operational expense owing to significant drop in internal cell resistance. Nonetheless, EAOPs are still prone to electrode fouling due to the occurrence of local alkalization on cathode surface. The issue worsens while operating at industrial scale where the electrochemical cells are operated in continuous mode. Once the cathode is totally passivated by the scaling, the EAOPs efficacy is undoubtedly compromised. Polarity reversal has been a common practice in industries to periodically redissolve the deposition. Both cathode and anode materials need to be stable against corrosion/burning during the oxidation period, consequently the cathode material is often expensive. Therefore, in quest for limiting or even avoiding the polarity reversal, the objective of this thesis was to identify the optimal operating conditions by which minimal electro-precipitation was formed during an electro-oxidation of effluent to maximize the process efficacy. Decisive parameters on the kinetics of electroprecipitation were scrutinized, i.e., the applied current density  $(j_{app})$ , the interelectrode distance  $(d_{elec})$ , the matrix of effluent to be treated and the cathode material.

Under the operating conditions applied, only Mg(OH)<sub>2</sub>, CaCO<sub>3</sub> and Ca<sub>3</sub>(PO<sub>4</sub>)<sub>2</sub> precipitates were present. Other deposits were not thermodynamically favorable. Mg(OH)<sub>2</sub> precipitation was highly dependent on  $j_{app}$  whilst CaCO<sub>3</sub> and Ca<sub>3</sub>(PO<sub>4</sub>)<sub>2</sub> easily occurred even at low applied current. Then, more scaling was observed in submillimetric range due to the involvement of more vigorous concomitant hydrogen gas evolution in macrometric setup. A mathematical model was proposed to predict electro-precipitation at varying  $d_{elec}$ , which account for scaling detachment due to the involvement of gas activity. The model would better allow estimating the rate of mineral scaling and could as well optimize the polarity reversal frequency. When the comparison was made with real wastewater, more electro-precipitation occurred due to co-precipitation of dissolved organics and inhibiting effect of sulfate in simulated effluent. When higher  $j_{app}$  was adopted, more organic mineralization, higher target pharmaceutical (tylosin) degradation and less mineral scaling were noticed. This was at the expense of higher power consumption and significantly higher release of undesirable byproducts. Finally, the use of porous carbonaceous cathodes only yielded in more electro-precipitation.

In perspective, the use of porous cathode in conjunction with micrometric  $d_{\rm elec}$  would benefit from efficient electro-precipitation for value-added non-conducting mineral resources recovery. It could be implemented alongside EAOP to achieve simultaneous pollutant degradation and resource recovery in a single operation.

### RÉSUMÉ

Les procédés électrochimiques d'oxydation avancés (PEOAs) sont de plus en plus reconnus pour éliminer efficacement les polluants organiques, y compris les micropolluants, dans les eaux usées. Les radicaux hydroxyles en tant qu'oxydants puissants et très peu sélectifs y sont générés in situ et en continu. L'application plus récente de réacteurs à couche mince dans le cadre des PEOAs a permis d'améliorer l'efficacité de traitement d'effluents aqueux, suite à l'intensification du transport de matière et des cinétiques de réactions d'oxydoréduction. A cela s'ajoute une baisse des coûts opérationnels grâce à la réduction significative de la résistance interne des cellules. Néanmoins, les PEOAs restent susceptibles à l'encrassement dû à l'alcalinisation locale sur la surface de la cathode. La problématique s'aggrave lors du fonctionnement à plus grande échelle, où les cellules s'opèrent en mode continu. Or, une fois que la cathode est complètement passivée par les dépôts minéraux, l'efficacité de l'oxydation devrait être compromise. En pratique, l'inversion de polarité est souvent plébiscitée en électrolyse industrielle pour redissoudre périodiquement les précipités. Par conséquent, les matériaux de cathode et d'anode doivent être stables contre la corrosion/l'incinération durant la période d'oxydation, et cela implique l'utilisation de matériaux de cathode coûteux. Afin de limiter, voire d'éviter cette inversion de polarité, l'objectif principal de cette thèse a été d'identifier les conditions opératoires optimales pour lesquelles l'entartrage soit minimal et la durée d'efficacité d'oxydation anodique maximale. Les paramètres impactant la cinétique d'électro-précipitation ont été examinés, tels que la densité de courant appliquée  $(j_{app})$ , la distance interélectrode  $(d_{elec})$ , la matrice de l'effluent à traiter et le matériau de cathode.

Dans les conditions de fonctionnement appliquées, seuls les précipités de  $Mg(OH)_2$ ,  $CaCO_3$  et  $Ca_3(PO_4)_2$  étaient présents, les autres n'étant pas thermodynamiquement favorisés. L'électro-précipitation de  $Mg(OH)_2$  dépendait fortement de  $j_{app}$  tandis que les précipités de  $CaCO_3$  et de  $Ca_3(PO_4)_2$  se produisaient facilement, même à faible courant. De plus, un dépôt plus important a été observé dans la condition microfluidique due à l'évolution concomitante plus vigoureuse de l'hydrogène gazeux dans la configuration millimétrique. Un modèle mathématique a été proposé pour prédire l'électro-précipitation aux différentes  $d_{elec}$ , en tenant compte du détachement des précipités lié à cette évolution de gaz. Ce modèle devrait permettre à l'avenir de pouvoir mieux estimer les rendements d'électro-précipitation tout en optimisant la fréquence d'inversion de polarité. De surcroît, lorsque la comparaison a été faite en utilisant des eaux usées traitées municipales, plus d'encrassement s'est produit avec la matrice réelle, en raison de la co-précipitation des matières organiques dissoutes et de l'effet inhibiteur des sulfates dans l'effluent simulé. Lorsque qu'une  $j_{app}$  élevée a été adoptée, les cinétiques de dégradation et de minéralisation du polluant pharmaceutique ciblé (tylosin) ont été augmentées et la quantité de dépôt cathodique diminuée. Cependant, cela s'est

accompagné d'une consommation énergétique significativement plus élevée et d'un relargage de sous-produits indésirables plus conséquent auquel il faudra remédier dans le futur. Enfin, l'utilisation de cathodes plus poreuses n'a fait qu'accroître le phénomène d'électro-précipitation.

En perspective, la porosité élevée de la cathode accompagnée d'une faible  $d_{\rm elec}$  semblent être des conditions importantes dans un objectif de récupération cathodique optimale de minéraux non conducteurs à haute valeur ajoutée. Cette électro-précipitation pourrait se réaliser en complément des PEOAs qui permettraient d'éliminer en parallèle les composés organiques dans l'effluent.

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with 10 mg L <sup>-1</sup> tylosin at (a) 0.8, (c) 4 and (e) 16 mA cm <sup>-2</sup> as well as the distribution of N-species at (b) 0.8,
(d) 4 and (f) 16 mA cm <sup>-2</sup> . (□): Cl <sup>-</sup> , (□): HClO/ClO <sup>-</sup> , (□): ClO <sub>2</sub> <sup>-</sup> , (□): ClO <sub>4</sub> <sup>-</sup> and (□): Cl-
volatile, while ( $\Delta$ ): NH <sub>4</sub> <sup>+</sup> , ( $+$ ): NO <sub>2</sub> <sup>-</sup> , ( $\square$ ): NO <sub>3</sub> <sup>-</sup> , ( $\square$ ): N-volatile and ( $\times$ ): N-combined. Anode: BDD and
cathode: stainless steel
Fig. VII.12. Kinetics and modeling of (a) tylosin degradation, (b) zoom into the early kinetics of tylosin
degradation and (c) mineralization during the electro-oxidation of RW containing 10 mg $L^{-1}$ tylosin. (O, —):
0.8, $(\times, -)$ : 4 and $(\Delta, -)$ : 16 mA cm <sup>-2</sup> current densities. $(\Box, -)$ : simulated wastewater (SW) spiked with
10 mg L <sup>-1</sup> tylosin (4 mA cm <sup>-2</sup> ). Anode: BDD and cathode: stainless steel

<b>Fig. VIII.1</b> . Voltammograms on (a) carbon paper, (b) graphite and (c) stainless steel for the determination of electroactive surface area. Electrolyte: 500 mL of 50 mM of $K_4Fe(CN)_6$ in 0.50 M of $Na_2CO_3$ ; potential sweep between 0.6 to -1.6 V/Ag-AgCl with a scan rate of 10 mV s <sup>-1</sup> ; CE: BDD and RE: Ag-AgCl; $d_{elec}$ : 500
μm
10 mV s <sup>-1</sup> ; CE: BDD and RE: Ag-AgCl; $d_{\text{elec}}$ : 500 $\mu$ m.
Fig. VIII.3. Photographs of mineral electro-precipitation occurring on (a) stainless steel, (b) graphite and (c,
<b>d</b> ) carbon paper cathodes after 5 h electrolysis. $d_{\text{elec}}$ : 500 µm. $I_{\text{app}}$ : 200 mA
Fig. VIII.4. Concentrations of Mg <sup>2+</sup> , Ca <sup>2+</sup> and TIC during the electrolysis at 20 mA ((a), (c), (e)) and 200
$mA$ (( <b>b</b> ), ( <b>d</b> ), ( <b>f</b> )). $d_{elec}$ : 500 μm, ( <b>X</b> ): stainless steel, ( <b>O</b> ): graphite and ( $\square$ ): carbon paper cathode.
Electrolyte initially contained 150 mg L <sup>-1</sup> Ca <sup>2+</sup> , 5 mg L <sup>-1</sup> Mg <sup>2+</sup> and 60 mg-C L <sup>-1</sup> of TIC
Fig. VIII.5. Evolution of bulk pH during the electrolysis at (a) 20 mA and (b) 200 mA. Cathode material:
$(\mathbf{X})$ : stainless steel, $(\mathbf{O})$ : graphite and $(\square)$ : carbon paper
Fig. VIII.6. Concentrations of (a) Mg <sup>2+</sup> , (b) Ca <sup>2+</sup> , (c) TIC, (d) effluent bulk pH, (e) PO <sub>4</sub> <sup>3-</sup> and (f) TOC during
the electro-oxidation treatment of RW at 200 mA applied current on (X) stainless steel and (O) carbon paper
cathode. The properties of RW are given in <b>Table VII.1</b> of <b>Section VII.2</b>
Fig. VIII.7. Concentrations of (a) NO <sub>3</sub> -, (b) NH <sub>4</sub> +, (c) HClO/ClO-, (d) Cl-, (e) ClO <sub>3</sub> - and (f) ClO <sub>4</sub> - during the
electro-oxidation treatment of RW at 200 mA applied current on (X) stainless steel and (O) carbon paper
cathode. The properties of RW are given in <b>Table VII.1</b> of <b>Section VII.2</b>
Fig. VIII.8. SEM of unused carbon paper (control) ((a), (b)) and electro-precipitate formed on carbon paper
after 5 h electrolysis of BE ((c), (d)) and RW ((e), (f)). $I_{app}$ : 200 mA
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initially containing 150 mg L <sup>-1</sup> Ca <sup>2+</sup> , 5 mg L <sup>-1</sup> Mg <sup>2+</sup> and 60 mg-C L <sup>-1</sup> of TIC at 200 mA. $d_{\text{elec}}$ of $(\Delta, \blacktriangle)$ : 100,
$(O, \bullet)$ : 500 and $(\Box, \blacksquare)$ : 3000 $\mu m$ were evaluated using $(\blacktriangle, \bullet, \blacksquare)$ : graphite and $(\Delta, O, \Box)$ : stainless steel
cathode. Anode: BDD
Fig. VIII.10. LSV of stainless steel and graphite plate cathodes when they were assembled in an
electrochemical cell with (a) 500, (b) 3000 and (c) 100 $\mu$ m $d_{\text{elec}}$ . The potential was swept between 0.5 to -1.8
V/Ag-AgCl with a scan rate of 10 mV s <sup>-1</sup> . CE: BDD, RE: Ag-AgCl
Fig. VIII.11. Evolutions of $R_{\text{CT(G)}}(\triangle, \bullet, \blacksquare)$ and $R_{\text{CT(G)}}/R_{\text{CT(SS)}}(\square)$ ((a), (c), (e)) as well as $C_{\text{DL(G)}}(\triangle, \bullet, \blacksquare)$
$\blacksquare$ ) and $C_{DL(G)}/C_{DL(SS)}$ ( $\square$ ) (( <b>b</b> ), ( <b>d</b> ), ( <b>f</b> )) during the electrolysis of solution containing Ca <sup>2+</sup> 150 mg L <sup>-1</sup> , Mg <sup>2+</sup>
5 mg L <sup>-1</sup> and TIC 60 mg L <sup>-1</sup> . ( <b>Δ</b> ): 100 μm (( <b>a</b> ), ( <b>b</b> )), ( <b>Φ</b> ): 500 μm (( <b>c</b> ), ( <b>d</b> )) and ( <b>■</b> ): 3000 μm (( <b>e</b> ), ( <b>f</b> )). G:
graphite and SS: stainless steel
Fig. VIII.12. Evolutions of $R_F$ during the electrolysis of solution initially containing $Ca^{2+}$ 150 mg $L^{-1}$ , $Mg^{2+}$ 5
mg L <sup>-1</sup> and TIC 60 mg L <sup>-1</sup> . (♠): 500 and (♠): 3000 μm. WE: graphite, CE: BDD and RE: Ag-AgCl 250
<b>Fig. IX.1</b> . Variation of Mg(OH) <sub>2</sub> and CaCO <sub>3</sub> electro-precipitation, cell voltage ( $\Delta U$ ), power consumption
(PC) and specific power consumption per gram of precipitate (SPCP) as a function of $d_{\text{elec}}$ . The reported
value of a given parameter was relative to the highest one obtained with a given $d_{\text{elec}}$ ( <b>Table IX.1</b> )
<b>Fig. IX.2</b> . Relative performances of the electro-oxidation of reclaimed wastewater spiked with 10 mg L <sup>-1</sup>
tylosin at 0.8, 4 and 16 mA cm <sup>-2</sup> during 5 h of treatment

## **ABBREVIATIONS**

Symbol	Unit	Definition	Symbol	Unit	Definition
α	-	OH- hindrance factor	η	%	Current efficiency
A	$m^2$	Electroactive surface area	$\eta_{\mathrm{O}_2}$	-	Efficiency of OH <sup>-</sup> production from dissolved ORR
$A_{ m C}$	$m^2$	Cross-sectional area	$\eta_{ m H_2}$	-	Loss of faradaic current due to concomitant HER
Ag-AgCl	-	Saturated silver chloride reference electrode	$\eta_{\rm H_2O}$	-	Efficiency of OH <sup>-</sup> production from dissolved HER
BDD	-	Boron-doped diamond	$I_{ m P}$	A	Peak current
BE	-	Basic effluent	IOA	-	Index of agreement
γ	$V s^{-1}$	Scan rate	I	A	Applied current intensity
$\Delta C$	mol m <sup>-3</sup>	Difference in concentration	$I_{ m lim}$	A	Limiting current
$C_{\mathrm{i}}$	mol m <sup>-3</sup>	Concentration of ionic species in electrolyte	$j_{ m app}$	A m <sup>-2</sup>	Applied current density
$C_{ m DL}$	F	Double layer capacitance	$\dot{J}_{ m lim,0}$	$A m^{-2}$	Initial limiting current density
CE	-	Counter electrode	κ	μS cm <sup>-1</sup>	Ionic conductivity containing dispersed gas bubble
$C_{ m F}$	F	Film capacitance	$\kappa_0$	μS cm <sup>-1</sup>	Ionic conductivity free of gas bubble
CPE	F	Constant phase element	$\kappa/\kappa_0$	-	Electrolyte reduced conductivity
$C_{\mathrm{SOL}}$	mol m <sup>-3</sup>	Concentration of reactive species in the bulk	$K_{\mathrm{Ai}}$	-	Acidity constant
Cl-volatile	mol m <sup>-3</sup>	Volatile chlorinated species consisted of Cl <sub>2</sub> , ClO <sub>2</sub> and Cl <sub>2</sub> O	$k_{CaCO_3}$	m <sup>4</sup> mol <sup>-1</sup> s <sup>-1</sup>	Heterogeneous second rate constant of CaCO <sub>3</sub> electroprecipitation
$[\mathcal{C}\mathcal{O}_3^{2-}]_{\mathrm{int}}$	mol m <sup>-3</sup>	Interfacial CO <sub>3</sub> <sup>2-</sup> concentration	$k_{Ca(OH)_2}$	$m^7 \text{ mol}^{-2} \text{ s}^{-1}$	Heterogeneous second rate constant of Ca(OH) <sub>2</sub> electroprecipitation
COD	$mg-O_2 L^{-1}$	Chemical oxygen demand	$k_{ m inhib}$	m <sup>4</sup> mol <sup>-1</sup> s <sup>-1</sup>	Coefficient of Mg <sup>2+</sup> inhibition by adsorption/incorporation
CV	-	Cyclic voltammetry	KKT	-	Kramers-Kronig Transform
d	m	Channel width (equivalent to W)	$k_{ m m}$	m s <sup>-1</sup>	Mass transfer coefficient
DAC	-	Dimensionless applied current density	$k_{ m MgCO_3}$	$m^4 \text{ mol}^{-1} \text{ s}^{-1}$	Heterogeneous second rate constant of MgCO <sub>3</sub> electro- precipitation Heterogeneous second rate
$d_{ m DL}$	m	Double layer thickness	$k_{\mathrm{Mg(OH)_2}}$	$m^7 \text{ mol}^{-2} \text{ s}^{-1}$	constant of Mg(OH) <sub>2</sub> electro- precipitation
$d_{ m elec}$	m	Interelectrode distance	$K_{\mathrm{S,i}}$	-	Solubility constant
$d_{ m H}$	m	Hydraulic diameter	$K_{ m W}$	-	Water auto-ionization constant
$D_{ m L}$	$m^2 s^{-1}$	Electrolyte diffusion coefficient	$\lambda_{i}$	$S m^2 mol^{-1}$	Ionic molar conductivity
$d_{ m elec}/W$	-	Aspect ratio	L	m	Length of electrode
$\mathcal{E}_0$	F.m	Vacuum permittivity	Le	-	Dimensionless distance
$arepsilon_{ ext{S}}$	-	Solvent dielectric constant	$L_{ m IND}$	Н	Inductance
$E_{ m A}$	V	Measured anode potential	LSV	-	Linear scan voltammetry
$E_{ m C}$	V	Measured cathode potential	$m_{\rm i}$	g	Mass of element
EEC	-	Electrical equivalent circuit	ME	-	Model efficiency
$E^0_{ m Oxy/Red}$	V	Standard potential of a redox species	$\nu_{\rm H_2}$	-	Stoichiometry of H <sub>2</sub> in HER
$E_{ m L}$	m	Entry length	$\nu_{\rm H_2O}$	-	Stoichiometry of H <sub>2</sub> O in HER

Symbol	Unit	Definition	Symbol	Unit	Definition
$ u_{0_2}$	-	Stoichiometry of dissolved O <sub>2</sub> in ORR	t	S	Electrolysis time
n	-	CPE parameter	$t_{ m i}$	=	Ionic transference number
$N_{ m i}$	mol	Molar quantity	TIC	mg-C L-1	Total inorganic carbon
$n_{ m e}$	-	Number of electrons participated in considered reaction	TOC	mg-C L <sup>-1</sup>	Total organic carbon
$n_{ m H_2}$	_	Number of e <sup>-</sup> in HER	TN	mg-N L <sup>-1</sup>	Total nitrogen
$n_{\mathrm{H_2O}}$	_	Number of e <sup>-</sup> in HER	$\Delta U$	V	Cell voltage
$n_{0_2}$	_	Number of e <sup>-</sup> in ORR	$u_{\rm i}$	$m^2 s^{-1} V^{-1}$	Ionic mobility
NOM	-	Natural organic matter	$u_{\rm L}$	m s <sup>-1</sup>	Cross-sectional electrolyte velocity
N-volatile	mol m <sup>-3</sup>	Volatile nitrogenous species	$v_{ m L}$	$m^2 s^{-1}$	Electrolyte kinematic viscosity
OER		consisted of N <sub>2</sub> and NO <sub>X</sub> O <sub>2</sub> evolution reaction	V	$m^3$	Treated volume
ORR	-	O <sub>2</sub> evolution reaction O <sub>2</sub> reduction reaction		$m^3 s^{-1}$	Volume flow rate
	mol m <sup>-3</sup>		$\dot{V}_{ m F}$	m <sup>3</sup>	
OH <sub>crit</sub>		Critical OH <sup>-</sup> concentration	$V_{ m R}$		Reactor volume
PC	kWh m <sup>-3</sup>	Specific power consumption	$rac{arphi}{W}$	-	Gas void Width of electrode
$\mathrm{pH}_{\mathrm{crit}}$ Pt	-	Critical pH Platinum	W WE	m	
	kg m <sup>-3</sup>		WE WWTP	-	Working electrode Wastawater treatment plant
$\rho_{\rm L}$	kg III	Electrolyte density Electrolyte relative resistivity	X	- %	Wastewater treatment plant Conversion rate
$ ho/ ho_0$	_	Electroryte relative resistivity	Λ	70	Electrode length (equivalent to
R	-	Gas constant	$x_{\rm e}$	m	L)
$\sum RI$	V	Internal ohmic resistance	$z_{i}$	-	Ionic valence number
Re	-	Reynolds number	ω	-	Constant relating $C_{\rm DL}$ and $d_{\rm elec}$
$R_{\rm CT}$	Ω	Charge transfer resistance			
$R_{ m DES}$	Ω	Adsorption/desorption resistance			
$R_{ m F}$	$\Omega$	Film resistance			
$r_{ m i}$	mol m <sup>-2</sup> s <sup>-1</sup>	Heterogeneous rate of reaction			
RE	_	Reference electrode			
RMSE	_	Root mean square error			
$R_{ m SOL}$	Ω	Electronic transfer resistance in			
DW		solution between WE and RE			
RW	$m^2$	Reclaimed wastewater			
S Sc	111	Electrode surface area Schmidt number			
SCE	-	Saturated calomel electrode			
SCE SD	-				
SHE	-	Supersaturation degree Standard hydrogen potential			
SHE SI	-	Supersaturation index			
SE	-	Saline effluent			
SPC	kWh g <sup>-1</sup>				
SPC	•	Specific power consumption			
SPCM	kWh g- TOC <sup>-1</sup>	Specific power consumption per gram of TOC			
SPCP	kWh g- precipitate <sup>-1</sup>	Specific power consumption per gram of electro-precipitate			
SPCT	kWh g-	Specific power consumption per			
	tylosin <sup>-1</sup>	gram of tylosin			
SSE	-	Saturated sulfate electrode			
SW	-	Simulated wastewater			
τ	S	Electrolyte resistance time			
T	K	Temperature			

## **RÉSUMÉ ÉTENDU**

#### I. Introduction générale

Ce travail de thèse a pour objectif la caractérisation et la quantification de l'électro-précipitation minérale cathodique ayant lieu au cours d'un traitement d'électro-oxydation d'eaux usées. Le procédé d'électro-oxydation a été appliqué dans un réacteur électrochimique submillimétrique ayant une surface d'électrode de 50 cm². La conception du réacteur submillimétrique, ou également appelé à couche mince, a été choisie en raison de l'énorme potentiel qu'il pourrait offrir pour des applications environnementales [1, 2]. Par exemple, le transport de matière et les réactions d'oxydoréduction sont intensifiés et les coûts opérationnels diminués grâce à une forte réduction de la résistance interne de la cellule [3-5]. En dépit de cet intérêt potentiel, peu de recherches ont été dirigées vers ce sujet. Plus précisément, aucune publication n'existe dans la littérature au regard de la formation d'électro-précipités dans une cellule électrochimique à couche mince, contrairement à celles faites dans le montage conventionnel à disque rotatif [6-8]. Une fois que la cathode est totalement passivée par l'électro-précipitation, l'efficacité de l'électro-oxydation est sans aucun doute compromise. C'est la raison pour laquelle les travaux de recherche de la thèse ont été menés sur cette problématique. La Fig. R.I illustre l'approche et le déroulement de la thèse.

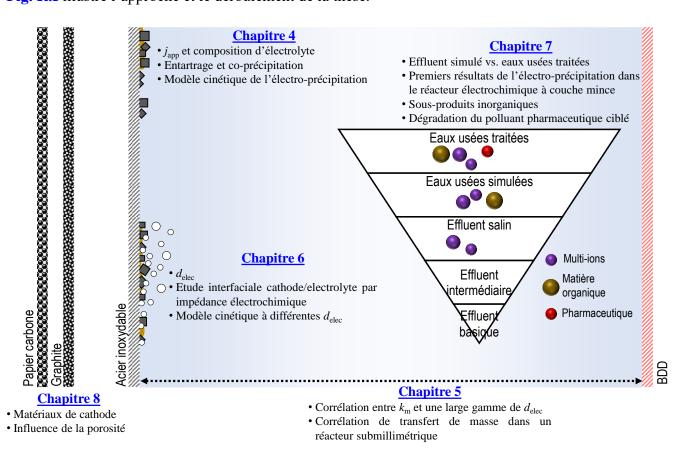


Fig. R.I. Schéma de l'approche et du déroulement de la thèse.

Selon la **Fig. R.I**, les paramètres déterminants sur la cinétique de l'électro-précipitation ont été examinés, tels que la densité de courant appliquée  $(j_{app})$ , la distance inter-électrode  $(d_{elec})$ , la matrice de l'effluent à traiter et le matériau de la cathode. Des modèles théoriques ont été proposés accompagnés des résultats expérimentaux. Du point de vue pratique, l'objectif a été d'établir les conditions opératoires pour lesquelles l'entartrage serait minimisé. Dans ce cas, l'impact sur l'efficacité de l'électro-oxydation serait minimisé et donc la durabilité du procédé serait assurée.

#### II. Influence de la densité de courant appliquée $(j_{app})$

Dans la gamme de  $j_{app}$  utilisée, l'alcalinisation locale sur la surface de la cathode s'est produite principalement par la réduction de l'eau au lieu de celle de l'O<sub>2</sub> dissous. L'entartrage était indépendant du phénomène d'électro-migration et était contrôlé par la diffusion du transport de matière. Dans toutes les conditions étudiées, seuls les précipités d'hydroxyde de magnésium (Mg(OH)<sub>2</sub>), de carbonate de calcium (CaCO<sub>3</sub>) et de phosphate de calcium (Ca<sub>3</sub>(PO<sub>4</sub>)<sub>2</sub>) ont été formés. La formation d'hydroxyde de calcium (Ca(OH)<sub>2</sub>), de carbonate de magnésium (MgCO<sub>3</sub>), de struvite (MgNH<sub>4</sub>H<sub>n</sub>PO<sub>4</sub><sup>3-n</sup>), de dolomite (CaMg(CO<sub>3</sub>)<sub>2</sub>) et de phosphate de magnésium (Mg<sub>3</sub>(PO<sub>4</sub>)<sub>2</sub>) n'était pas favorisée. L'électro-précipitation de Mg(OH)<sub>2</sub> s'est avérée fortement dépendante de  $j_{app}$ , mais indépendante des autres espèces ioniques coprésentes dans l'électrolyte. Mg(OH)<sub>2</sub> précipitait dès que le pH interfacial de 10,2 était atteint. Le CaCO<sub>3</sub> se déposait facilement sur la surface de la cathode même sous une faible polarisation du courant. L'entartrage par CaCO<sub>3</sub> était plus important à faible  $j_{app}$  en raison de l'évolution concomitante du gaz plus intense à  $j_{app}$  élevée. Ensuite, les études ont été dirigées vers l'électro-précipitation au sein des réacteurs électrochimiques avec des  $d_{elec}$  variables.

# III. Evolution de transport de matière d'une échelle macrométrique à l'échelle micrométrique

L'évolution du transport de matière en fonction de  $d_{\rm elec}$  pour des variations de ce paramètre allant d'une échelle micrométrique (100 µm) à millimétrique (3000 µm) a été préalablement examinée et a été représentée dans la **Fig. R.II**. Un changement drastique de l'évolution de  $k_{\rm m}$  a permis d'identifier de manière quantitative le début du comportement microfluidique, qui se situe en dessous de 1000 µm. Une corrélation a été proposée pour prédire  $k_{\rm m}$  en fonction de la large gamme de  $d_{\rm elec}$ . En écoulement laminaire (7 < Reynolds < 623) et pour une température comprise entre 10 et 50°C (532 < Schmidt < 3315), de nouvelles corrélations de transport de matière utilisant le nombre adimensionnel de Sherwood ont été obtenues pour les configurations microfluidiques et millimétriques. L'extrapolation de  $k_{\rm m}$  pour la mise à l'échelle des réacteurs électrochimiques microfluidiques parait donc réalisable.

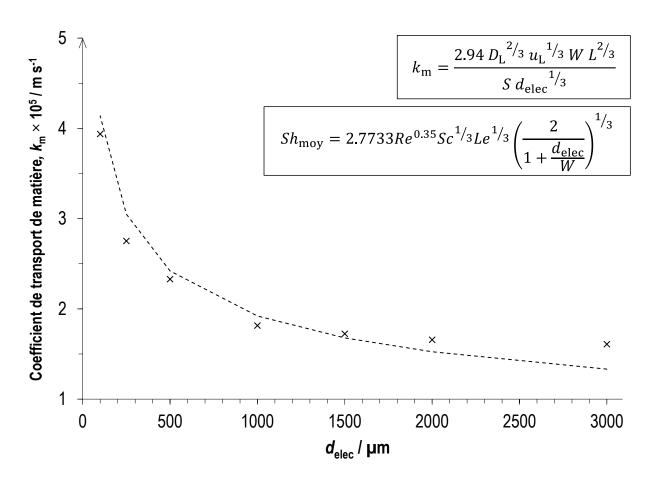


Fig. R.II. Corrélation de  $k_{\rm m}$  en fonction de  $d_{\rm elec}$  et corrélation obtenue avec le nombre adimensionnel de Sherwood.

## IV. Rôle de la distance inter-électrode ( $d_{\text{elec}}$ ) sur la formation d'électroprécipitation

Une fois la caractérisation du transport de matière en fonction de  $d_{\rm elec}$  appréhendée, la recherche s'est orientée vers le rôle de  $d_{\rm elec}$  sur la formation des électro-précipités dans la cellule électrochimique. Comme cela a été prédit par un nouveau modèle mathématique prenant en compte la réaction concomitante d'évolution de l'hydrogène (H<sub>2</sub>) (HER), une précipitation plus intense a été observée par ordre décroissant de  $d_{\rm elec}$  de 3 mm à 100 µm, en appliquant 4 mA cm<sup>-2</sup>. En revanche, aucun dépôt n'était présent à la plus faible  $d_{\rm elec}$  étudiée (50 µm), ce qui est dû à une condition non faradique. Le potentiel cathodique appliqué ( $E_{\rm C}$ ) a diminué avec l'augmentation de la  $d_{\rm elec}$ , ce qui a intensifié la production de bulles de H<sub>2</sub> et minimisé l'électro-précipitation. Des études supplémentaires en appliquant un  $E_{\rm C}$  identique ont mis en évidence la contribution anodique de l'oxygène (O<sub>2</sub>) dissous produit à l'anode sur la cinétique d'électro-précipitation. Cet apport est encore plus intensifié à des distances submillimétriques. La concomitance de HER en fonction de  $d_{\rm elec}$  a été évaluée plus en détail en accédant à la résistivité électrolytique relative ( $\rho/\rho_0$ ) et à la fraction du gaz ( $\varphi$ ) formé pendant l'électrolyse. Les résultats montrent que  $\rho/\rho_0$  et  $\varphi$  augmentaient avec la diminution de  $d_{\rm elec}$ . Malgré

l'évolution de HER en parallèle, lorsque  $d_{\rm elec}$  diminuait, la résistance interne de la cellule électrochimique était réduite. Elle a diminué linéairement jusqu'à un  $d_{\rm elec}$  de 100  $\mu$ m, valeur en dessous de laquelle la linéarité a été interrompue. Enfin, une étude d'impédance électrochimique a été réalisée pour caractériser les processus électrochimiques à l'interface cathode/électrolyte. De nouveaux modèles mathématiques corrélant la résistance de transfert de charge ( $R_{\rm CT}$ ) et la capacitance de double couche électronique ( $C_{\rm DL}$ ) en fonction de  $d_{\rm elec}$  ont été proposés. L'interrelation entre ces paramètres fondamentaux avec l'occurrence d'électro-précipitation est schématisée dans la **Fig. R.III**.

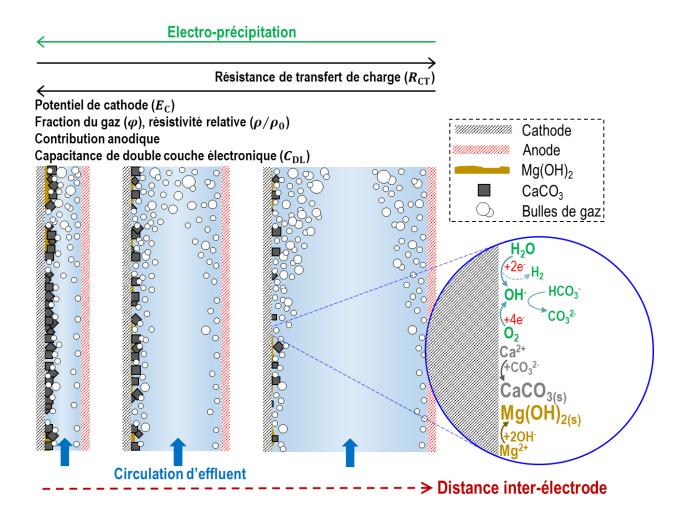


Fig. R.III. Schéma mécanistique de la formation d'électro-précipités sous l'influence de l'évolution concomitante du  $H_2$  en fonction de  $d_{\text{elec}}$ .

#### V. Rôle de la matrice d'effluent : effluent simulé vs. eaux usées traitées

La thèse s'est poursuivie sur des études inédites concernant le rôle de la matrice de l'effluent sur l'électro-précipitation dans le réacteur électrochimique à couche mince. L'augmentation du degré de complexité de la matrice depuis l'effluent simulé jusqu'au système réel constitué d'eaux usées urbaines traitées a été abordée. En complément des phénomènes d'entartrage, la minéralisation de l'effluent, le

devenir des sous-produits inorganiques ainsi que la dégradation d'un polluant pharmaceutique représentatif (tylosine) ont été suivis. Il a été démontré dans des eaux simulées que la présence d'ions multiples (Ca²+, Mg²+, HCO₃-/CO₃²-, NH₄+, H₂PO₄-/HPO₄²-, NO₃-, NO₂-, Cl⁻, SO₄²-, K+ et Na+) et de matière organique (peptone et extrait de viande) représentatifs de l'effluent réel ont peu d'influence sur la cinétique de formation d'électro-précipitation. Cette tendance a été attribuée à l'occurrence d'un pH local élevé sur la cathode qui pourrait atténuer l'effet d'inhibition des matières organiques sur la formation du dépôt [9]. Lorsque les dépôts ayant lieu dans l'effluent simulé sont comparés à ceux obtenus avec des eaux usées traitées, une quantité plus importante est observé dans ce dernier cas. Cela pourrait être dû, soit à la co-précipitation des matières organiques dissoutes - plus complexes dans les eaux usées réelles - avec Ca²+ et Mg²+ sur la surface de la cathode, soit à l'effet inhibiteur résultant de la teneur plus élevée en sulfates dans l'effluent simulé. Lors du traitement des eaux usées réelles, une minéralisation plus élevée est obtenue, mais elle s'accompagne d'une dégradation plus lente du polluant pharmaceutique ciblé. Cette dernière tendance est supposée liée à la compétition entre les substances organiques dissoutes dans l'effluent avec le polluant organique.

#### VI. Influence du matériau de la cathode et porosité

Enfin, l'influence des matériaux de cathode a été examinée. La cinétique et le mécanisme de formation d'électro-précipités sur l'acier inoxydable (non poreux) et des matériaux carbonés et poreux tel que le graphite et le papier carbone ont été comparés. Ils ont chacun une surface électroactive et une porosité différente. De plus, les propriétés interfaciales de ces trois cathodes, dans l'état initial ou après avoir subi un dépôt, ont été caractérisées par impédance électrochimique. Les résultats montrent que l'utilisation d'une cathode poreuse telle que le papier carbone entraînerait une précipitation plus rapide de CaCO<sub>3</sub>, un taux plus élevé de conversion de Cl<sup>-</sup> en sous-produits toxiques (ClO<sub>3</sub><sup>-</sup> et ClO<sub>4</sub><sup>-</sup>) ainsi qu'une acidification du milieu qui impliquerait un post-traitement. Par conséquent, ce n'était pas la cathode la plus appropriée pour répondre à l'objectif initial. L'électro-précipitation était légèrement plus favorisée sur le graphite que sur l'acier inox en raison du pH local plus élevé sur le premier. Par conséquent, même si le matériau carboné présente un avantage économique, son utilisation a entraîné plus d'électro-précipitation.

#### VIII. Conclusions et perspectives

En vue des résultats expérimentaux acquis au cours de la thèse, il a été démontré que l'acier inoxydable serait le matériau le plus adapté si la formation de précipités est à minimiser. Par ailleurs, les résultats suggèrent également que le papier carbone et d'autres matériaux poreux seraient les plus appropriés au cas où la précipitation cathodique est recherchée. Ils viendraient ainsi en prérequis du procédé d'électro-précipitation pour la récupération d'éléments à haute valeur ajoutée tel que le

phosphore. En termes de  $j_{app}$ , la valeur dans la gamme de 1 à 2 mA cm<sup>-2</sup> pourrait être un bon compromis tenant compte la consommation d'énergie globale, l'électro-précipité formé, la minéralisation de l'effluent ainsi que l'évolution des sous-produits indésirables. La durée totale du traitement devrait être plus longue pour maximiser le taux d'élimination de polluants, mais cela ne poserait pas de contrainte énergétique lorsque des réacteurs à couche mince seront adoptés. Enfin, les modèles cinétiques de formation de précipités développés dans la thèse pourraient être utiles pour prédire les phénomènes d'électro-précipitation. Ces modèles sont applicables soit pour limiter l'entartrage lors du procédé de traitement par l'électro-oxydation, ou bien pour favoriser le dépôt lorsqu'il est souhaité, ce qui est par exemple le cas dans le procédé de valorisation des déchets par l'électro-précipitation.

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#### **SCIENTIFIC PRODUCTION**

## Papers in peer-reviewed journals

- **1.** F.H. Adnan, E. Mousset, S. Pontvianne, M.-N. Pons, *Mineral cathodic electro-precipitation and its kinetic modelling in thin-film microfluidic reactor during advanced electro-oxidation process*, **Electrochimica Acta**, 387 (2021) 138487.
- **2.** F.H. Adnan, M.-N. Pons, E. Mousset, *Mass transport evolution in microfluidic thin film electrochemical reactors: New correlations from millimetric to submillimetric interelectrode distances*, **Electrochemistry Communications**, 130 (2021) 107097.
- **3.** F.H. Adnan, S. Pontvianne, M.-N. Pons, E. Mousset, Unprecedented roles of submillimetric interelectrode distances and electrogenerated gas bubbles on mineral cathodic electroprecipitation: modeling and interface studies, Chemical Engineering Journal, 431 (2022) 133413.

#### **Oral presentations**

- **1.** <u>F.H. Adnan</u>, M.-N. Pons, E. Mousset, Mineral scaling on the surface of electrode during electrochemical advanced oxidation process in microfluidic thin film reactor, 1<sup>st</sup> ELO. WatR international workshop, Nancy, France (15<sup>th</sup> 17<sup>th</sup> September 2020).
- 2. <u>F.H. Adnan</u>, M.-N. Pons, E. Mousset, Influence of micrometric interelectrode distances and cathode materials on mineral scaling during advanced oxidation process, 29<sup>th</sup> topical meeting ISE, Mikulov (online), Czech Republic (18<sup>th</sup> 21<sup>st</sup> April 2021).
- **3.** <u>F.H. Adnan, M.-N. Pons, E. Mousset, Influence of micrometric interelectrode distance and water hardness on mineral scaling at electrode surface during electrochemical advanced oxidation process, 12<sup>th</sup> ESEE, Leeuwarden (online), the Netherlands (14<sup>th</sup> 17<sup>th</sup> June 2021).</u>
- **4.** <u>F.H. Adnan, M.-N. Pons, E. Mousset, Influence of electro-precipitation on pharmaceutical micropollutant degradation in simulated and reclaimed wastewater, 72<sup>nd</sup> annual meeting ISE, Jeju Island (online), South Korea (29<sup>th</sup> 3<sup>rd</sup> September 2021).</u>
- 5. F.H. Adnan, M.-N. Pons, E. Mousset, Influence de la dureté des eaux usées et des microdistances sur l'efficacité d'un proceed électrochimique, 13<sup>th</sup> international conference GRUTTEE, Rennes, France (18<sup>th</sup> – 20<sup>th</sup> February 2020).

#### I. Introduction

Within the framework of water reuse as a countermeasure against water scarcity that might concern a magnitude number of countries across the globe, conventional biological and physicochemical treatments applied in wastewater treatment plants (WWTPs) do not suffice [1, 2]. In response, a variety of complementary treatments have been proposed across literature and some of them even have been applied at industrial scale [3-6]. The majority of them are based on the fundamentals of chemical oxidation, in which strong oxidants are involved in the degradation of pollutants [7, 8]. Classical processes such as chlorination, ozonation and advanced oxidation processes (AOPs) are to an extent well-established to be able to totally degrade organic contaminants present in aqueous media [3, 9]. Nevertheless, their efficiency comes first and foremost with high operational price. Secondly the addition of chemicals into the media to be treated is a prerequisite, be it H<sub>2</sub>O<sub>2</sub>, Cl<sub>2</sub>, O<sub>3</sub>, acid/base and/or ferrous sources. One might notice the contradictory practices between preserving the aquatic ecosystem receiving the treated wastewater against the admittance of unnatural element into it when chemical reagents are introduced into the media to be treated.

AOP techniques have been doped by the introduction of electrochemical advanced oxidation processes (EAOPs) [10-12]. The latter have been documented across literature to also be able to degrade organic pollutants into simpler organic compounds which are easier to be biodegraded [13]. A quasi-complete mineralization into CO<sub>2</sub> and inorganic species can be even obtained if required. A huge boost to the limitation of AOPs is that EAOPs do not necessarily require the addition of chemicals to electrogenerate strong, quasi non-selective oxidizing agent such as hydroxyl radicals (\*OH) as well as hydrogen peroxide (H<sub>2</sub>O<sub>2</sub>) *in-situ*.

In addition to the advanced electro-oxidation by OH mentioned above, other indirect electro-oxidation (also called mediated electro-oxidation) can take place with active oxidants (HClO/ClO<sup>-</sup>, Cl<sub>2</sub>, O<sub>3</sub>, etc) depending on the composition of electrolyte [14, 15]. It exists other variants of EAOPs such as those where photo-, solar photo- and sono- irradiation sources have been called upon in a way to optimize the processes [12, 16-18]. The coupling between EAOPs have also been explored to combine heterogeneous and homogeneous OH production to improve treatment capacity [9, 19, 20].

Nevertheless, the phenomenon of electrode fouling during electrolysis, and particularly in EAOPs, is an inherent issue to be solved. The formation of irreversible polymeric film on anode [14] as well as polymeric and/or inorganic scaling [21-23] on cathode can hamper the process efficiency. Consequently, their durability under continuous operation, which is the ultimate goal in real field applications, is massively impacted, when regular and costly maintenance are not foreseen.

In recent years, the application of submillimetric reactors within the framework of EAOPs has spurred the capability of treatment of wastewater effluent using these techniques. The micrometric interelectrode distance ( $d_{elec}$ ) significantly accommodates the limitation by mass transfer often encountered in macro-reactors. Furthermore, the redox reactions taking place on electrodes are intensified by bringing them very close to each other. While treating urban WWTPs effluents, their ionic conductivity (i.e., averagely 1000 µS cm<sup>-1</sup>) is apparently too low for conventional undivided reactor ( $d_{\text{elec}}$  between 1 and 4 cm). This low value of electrical conductivity limits the efficiency and applicability of EAOPs in various terms; firstly, the huge ohmic drop in-between electrodes requires an application of significantly larger operating currents thus larger overall power consumption. Secondly, the oxidation of pollutants will quickly be limited by mass transfer towards the electrodes. Thus, the efficiency of pollutant degradation is critically impacted. Thirdly, to promote the mass transfer of pollutants, supporting electrolyte is added making the EAOPs less attracting as environmental-friendly processes and therefore real field applications. Fortunately, the use of microfluidic reactors would allow the treatment of reclaimed wastewater without the need for supporting electrolyte. As a consequence, microfluidic configuration would allow not only innate mass transfer enhancement, but also huge cut in operation cost while being environmentally friendly.

Considering the interest in the use of submillimetric electrochemical reactor to treat wastewater, this topic deserves more attention from research community. Moreover, no work has yet been reported in literature pertaining the occurrence of mineral scaling inside a microfluidic electrochemical reactor. Consequently, the lack of data on scaling formation during an EAOP to treat either synthetic or real wastewater inside submillimetric as well as millimetric reactor configurations was deemed a crucial research gap. Hence, this PhD thesis work aims at evaluating the kinetics of electro-precipitation formation on cathode in scalable conventional parallel-plate reactor subjected to a parametric investigation. Primordial variables such as applied current density  $(j_{app})$ ,  $d_{elec}$ , matrices of effluent with increasing degree of complexity as well as cathode material were scrutinized. Theoretical predictive models have also been proposed accompanying the experimental results obtained. The mechanistic understanding on the formation of electro-precipitation at the cathode/electrolyte interface has been substantiated by means of electrochemical impedance studies. The practical and end objective is to establish the operating conditions by which less mineral scaling is formed, without impacting the EAOP efficiency to maximize the durability of the process.

The approach and structure of the thesis are schematized in Fig. I.1.

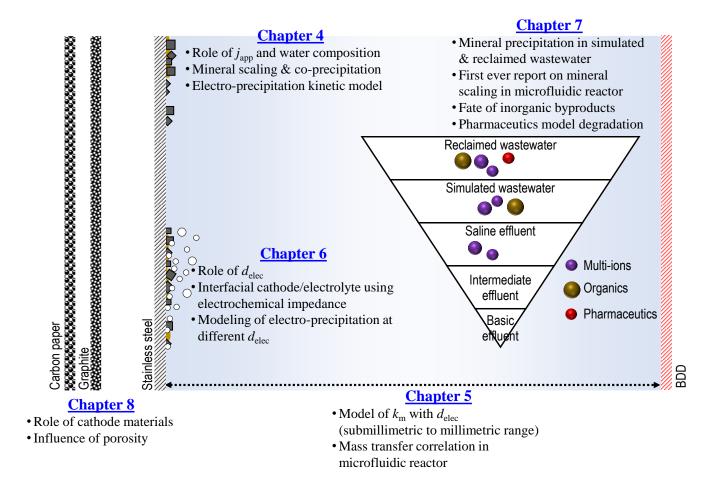


Fig. I.1. Thesis schematic chart and key parameters discussed in each chapter.

The manuscript starts with a bibliographical study in Chapter 2 which lays out the background on the opportunity surrounding the microfluidic technology towards its application in water treatment domain. The study also points out the lack of mass transfer correlation in such reactor design for easy scale up or scale down application. Furthermore, the application of electrochemical methods for calcareous scaling investigation, pollutant removal and recovery as well as water softening have been highlighted. It ends with a discussion on vast application of EAOP to treat wastewater of various sources. Chapter 3 details the technical aspects of experimental works including simulated and real wastewater effluent preparation, experimental pilot and electrochemical reactor setup, detailed analytical methods as well as modeling software. Next, the fundamental role of  $j_{app}$  and coprecipitation of mineral scaling in simulated effluent is discussed in Chapter 4. The simulated effluent mimicked the ionic properties of municipal reclaimed wastewater. The kinetics of electroprecipitation as function of  $j_{app}$  and the composition of effluent were depicted and modeled. Chapter 5 involves the study on the development of mass transfer correlation inside an electrochemical reactor operated with wide range of  $d_{\text{elec}}$ . The interelectrode gap ranged between submillimetric to millimetric values. A theoretical model between mass transfer coefficient  $(k_{\rm m})$  and the large spectrum of  $d_{\rm elec}$  is correlated. The established correlations are not only limited to water treatment domain but are also open to other various applications. Chapter 6 presents the critical role of  $d_{\text{elec}}$  towards the quantity

#### Chapter I

and kinetics of mineral scaling from microfluidic up to millimetric reactor scale. Determining parameters as the outcome of varying  $d_{\rm elec}$  were exposed. The underlying interfacial properties between partially blocked cathode surface in contact with the effluent was also discussed. The kinetics of electro-precipitation occurring under wide range of  $d_{\rm elec}$  were substantiated using mathematical models. In the next chapter, systematic and scrupulous comparison between the formation of mineral scaling in simulated and real reclaimed wastewater was undertaken in **Chapter 7**. Particularly, the influence of the occurring electro-precipitation on simultaneous evolution of organics mineralization, chlorinated and nitrogenous inorganic byproducts as well as on pharmaceutical pollutant removal is discussed. Finally, **Chapter 8** inspects the role of cathode materials and porosity towards the formation of mineral deposition during the electro-oxidation of both simulated and reclaimed wastewater. Pioneering work was conducted on the comparison between the kinetics of electro-precipitation formed on different cathode materials during the electro-oxidation process operating inside a microfluidic cell. The manuscript ends with conclusions and perspectives in **Chapter 9**.

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#### II. State of the art

#### Introduction

The literature review is outlined in the following order: (1) the application of microfluidic reactors to treat wastewater, (2) electro-precipitation on cathode during electrochemical processes and (3) removal of pharmaceutical pollutants in various matrix of effluent by EAOPs. The chapter is ended with concluding remarks and perspectives that will be partly addressed through this thesis work.

#### II.1. Electrochemical microreactors for wastewater treatment application

The use of microfluidic electrochemical reactors has been introduced since the early 1980s, but their application in the field of wastewater treatment is relatively recent (2010). Only a handful of papers have been found on this topic (**Table II.1**), understandably due to the question of productivity to treat large volumes of water with a microfluidic technology. Nevertheless, promising results have been reported and the possibility of scale-up has also been explored.

#### II.1.1. Influence of applied current density

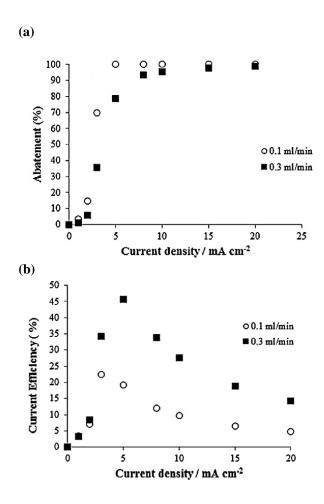
Applied current density is a crucial parameter in the application of electrochemical process to degrade organic pollutants. It is also closely related to the electrode potential through the Butler-Volmer relation [1], which is the driving force for direct and indirect oxidation and reduction reactions [2]. An optimal range of  $j_{app}$  was systematically noticed, depending on the type of EAOP [3-6], anode and cathode materials [5, 6], on operating conditions (Table II.1) as well as on the configuration mode of microfluidic reactors [5, 7, 8]. This apparent optimal value of  $j_{app}$  occurs due to two main reasons; either the electrical current is too low to produce significant quantity of oxidants or, on the contrary, too much electrical charges are given to the system till the redundant charges are wasted to unwanted side-reactions. Hence, the current efficiency ( $\eta$ , in %) is commonly written under the form given in Eq. (II.1) [5].

$$\eta = 100 \frac{n_e F C_{SOL} \dot{V}_F X}{j_{app} S} \tag{II.1}$$

where  $n_e$  is the number of required electrons for pollutant conversion, F is the Faraday constant (96485 C mol<sup>-1</sup>),  $\dot{V}_F$  is the flow rate (in m<sup>3</sup> s<sup>-1</sup>), X is the conversion rate,  $j_{app}$  in A m<sup>-2</sup>,  $C_{SOL}$  is the initial pollutant concentration (in mol m<sup>-3</sup>) and S is the electrode surface area (in m<sup>2</sup>).

**Figure II.1** illustrates the percentages and  $\eta$  of chloroacetic acid removal by electro-oxidation at different current densities and flow rates. From **Fig. II.1(a)**, it was observed that at low current

density, organic abatement increased with applied current. The degradation was kinetically controlled by charge transfer and that was the reason why it increased with  $j_{app}$ , whereas in the region of high  $j_{app}$ , the percentage of abatement increased negligibly or remained constant with further increase in  $j_{app}$ . The degradation was kinetically controlled by organics mass transfer to the electrode. Excessive charge would be used for parasitic reactions. As a result, the  $\eta$  plot depicted a maximum where it increased with further increase in  $j_{app}$  (Fig. II.1(b)). Several other works have come up with similar deduction [7-10].



**Fig. II.1**. (a) Percentage and (b) current efficiency  $(\eta)$  of degradation of chloroacetic acid in microfluidic reactor as a function of  $j_{app}$  at different flow rates. Initial pollutant concentration: 5 mM,  $d_{elec}$ : 50  $\mu$ m, anode: BDD, cathode: stainless steel, surface area: 5 cm<sup>2</sup>. Adapted from Scialdone et al. [5].

Finally, depending on how much electrical current is applied for the degradation of organics, specific energy consumption can be estimated for the process. Specific power consumption (*SPC*) has commonly been reported where the energy consumption was brought per gram of pollutant treated as defined in **Eq. II.2**.

$$SPC = \frac{\Delta UIt}{(\Delta C)V} \tag{II.2}$$

where SPC is generally in kWh per gram of pollutant,  $\Delta U$  is the cell voltage (in V), I is applied current intensity (in A), t is the time of electrolysis (in s),  $\Delta C$  is the decrease of pollutant concentration (in g L<sup>-1</sup>) and V is the treated volume in the reactor (in L).

SPC is typically proportional to the applied current. A large advantage of the microfluidic reactor in comparison with conventional device is that,  $\Delta U$  (Eq. (II.2)) is largely dropped owing to micrometric distance in-between electrodes [7, 8, 11, 12]. Consequently, SPC is significantly reduced hence the total EAOP operational cost could heavily be cut. Therefore, EAOP operating in microfluidic mode could then be competitive with respect to other conventional methods to treat wastewater, which is the reason why it has been scrutinized in this thesis.

#### II.1.2. Influence of flow rate

The effluent flow rate is an important parameter because it determines the residence time ( $\tau$ ) of the effluent to be treated in the electrolytic reactive zone. Generally, the influence of flow rate depends on the electrochemical process occurring on electrode surfaces. According to **Fig. II.1**, in the region where the process was kinetically controlled by charge transfer (lower  $j_{app}$ ), organic removal was more efficient with lower flow rate. It was due to higher  $\tau$ , thus there was a longer treatment of pollutant per charge applied. When the abatement of pollutant was controlled by mass transfer, the removal increased with the flow rate as a result of mass transfer intensification. *SPC* was also improved owing to similar reasons (**Fig. II.1**). Similar trend of conclusion has been documented in several works across literature [4-7, 9, 13].

#### II.1.3. Influence of interelectrode distance

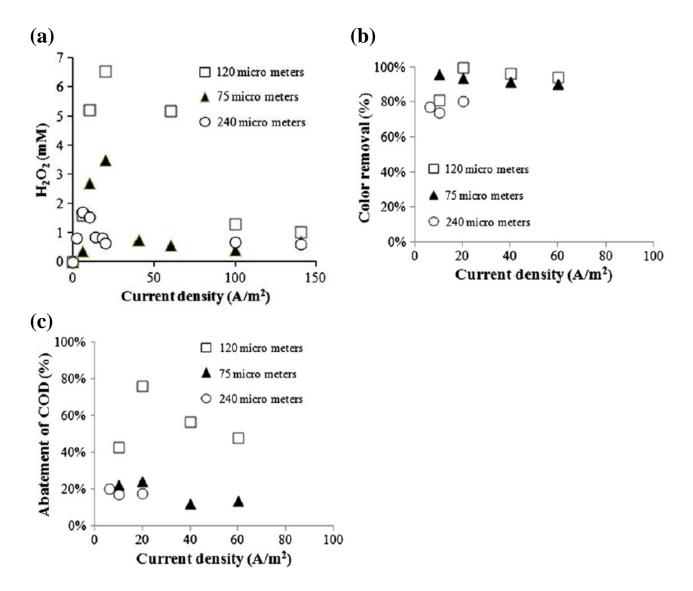
Despite being in micrometric conditions, the  $d_{\text{elec}}$  between anode and cathode still plays a determining role towards the reaction mechanism occurring inside the microfluidic reactor. **Figure II.2** plots the percentage of color and Chemical Oxygen Demand (COD) removal during an EF process to oxidize acid orange dye pollutant. The highest color (100%) (**Fig. II.2(b**)) and COD (80%) (**Fig. II.2(c**)) removals were achieved at lower  $d_{\text{elec}}$  (120 vs. 240 µm) at 20 A m<sup>-2</sup> (2 mA cm<sup>-2</sup>) [3]. It was attributed to higher H<sub>2</sub>O<sub>2</sub> production (**Fig. II.2(a**)) with smaller  $d_{\text{elec}}$  [3]. Lower distance resulted in a higher contribution of dissolved oxygen (O<sub>2</sub>) formed on anode via **Eq. (II.3)** to produce H<sub>2</sub>O<sub>2</sub> via **Eq. (II.4)**. Furthermore, the lower distance also led to a more uniform distribution of dissolved O<sub>2</sub> thus favoring H<sub>2</sub>O<sub>2</sub> formation [3]. Adversely, further decreasing in  $d_{\text{elec}}$  led to lower residence time in the reactor.

Thus, it led to lower accumulation of  $H_2O_2$  and faster mass transfer of  $H_2O_2$  towards the anode, before being oxidized back to dissolved  $O_2$  (Eq. (II.5)).

$$2H_2O \rightarrow O_2 + 4H^+ + 4e^-$$
 (II.3)

$$O_2 + 2H^+ + 2e^- \to H_2O_2$$
 (II.4)

$$H_2O_2 \to O_2 + 2H^+ + 2e^-$$
 (II.5)

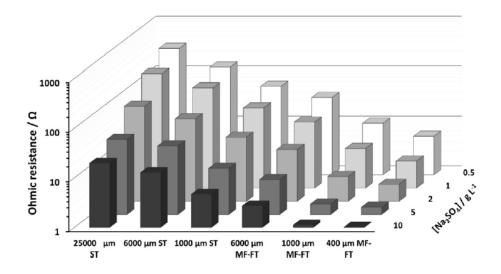


**Fig. II.2**. (a) Hydrogen peroxide (H<sub>2</sub>O<sub>2</sub>), percentage of (a) color removal and (b) COD removal of organic dye using EF inside microfluidic reactor with different  $d_{\text{elec}}$  at different  $j_{\text{app}}$ . Anode: IrO<sub>2</sub>-Ta<sub>2</sub>O<sub>5</sub>/Ti, cathode: graphite. Initial dye concentration: 0.43 mM, Fe<sup>2+</sup>: 0.5 mM. Adapted from Scialdone et al. [3].

When the interelectrode gap decreases, redox reactions on both anode and cathode are intensified. Mass transfer towards both electrodes is also significantly improved. This has been demonstrated during the electro-oxidation of tetrachloroethane inside a microfluidic reactor operating using 50 and

75  $\mu$ m  $d_{\text{elec}}$  [14]. Better abatement and  $\eta$  were noticed using lower  $d_{\text{elec}}$  and it was due to a better mass transfer of the pollutant towards the electrodes.

While progressively transitioning into the micrometric scale by reducing the interelectrode gap, the electrochemical cell voltage is reduced [7]. It occurs owing to the reduction of ohmic drop in-between electrodes. This phenomenon has clearly been presented in **Fig. II.3**. It illustrates the variation of ohmic resistance ( $\Sigma R$ ) happening inside electrochemical reactors with large range of  $d_{\rm elec}$  at different possible supporting electrolyte concentrations [8]. Considering for example the lowest supporting electrolyte (0.5 g L<sup>-1</sup>), the  $\Sigma R$  measured inside stirred tank reactor with 25000  $\mu$ m (2.5 cm)  $d_{\rm elec}$  was 346  $\Omega$ . The resistance was reduced to 6  $\Omega$  inside a microfluidic reactor operating using similar 0.5 g L<sup>-1</sup> supporting electrolyte.



**Fig. II.3**. Variation of ohmic resistance as a function of  $d_{\text{elec}}$  and different concentrations of supporting electrolyte. Adopted from Pérez et al. [8].

 $d_{\rm elec}$  plays a determining role on the kinetics of reactions inside microfluidic reactor. Therefore, its role on the formation of electro-precipitation taking place inside such reactor will be investigated in details in the thesis. There has been a study conducted by Khongton et al. in which the EO experiments were performed at different  $d_{\rm elec}$  (250 to 750  $\mu$ m), while keeping the  $\tau$  constant [9]. For that, the flow rate was adjusted according to the  $d_{\rm elec}$  to keep the residence time identical. Similar approach will be adopted throughout this work except when the influence of flow rate is intended.

#### II.1.4. Influence of electrode material

Depending upon the type of EAOP intended, electrode materials are selected accordingly. Advanced EO normally operates with anode possessing high overpotential for  $O_2$  evolution such as lead oxide (PbO<sub>2</sub>), tin oxide (SnO<sub>2</sub>) and BDD (boron-doped diamond), which possesses  $E_{H_2O/O_2}^0$  of 1.8-2.0, 1.9-

2.2 and 2.2-2.6 V/SHE respectively [15]. It allows the generation of potent oxidant such as \*OH on its surface via Eq. (II.6).

$$M + H_2O \rightarrow M(\bullet OH) + H^+ + e^-$$
 (II.6)

BDD has been receiving great attention due to its efficiency, stability, better life-span and high  $E^0_{\rm H_2O/O_2}$ . These anodes are called non-active anodes, because the produced  ${}^{\bullet}$ OH are loosely attached on anode surface by physisorption [16]. Hence, they can react quasi non-selectively with organic pollutants to oxidize them into simpler forms. When EO mode is solely intended, electrodes possessing high hydrogen (H<sub>2</sub>) evolution potential (close to 0 V/SHE) such as stainless steel, platinum (Pt), titanium (Ti), nickel (Ni) or other metals are used [17]. Thus, H<sub>2</sub>O<sub>2</sub> generation, which is as well an oxidant, is avoided.

When EF process is applied, cathode with low overpotential for  $H_2$  evolution (towards negative potential values) is used. The most common cathodes are carbonaceous-based materials such as graphite, carbon sponge, activated carbon fiber, graphite felt and carbon felt [15, 18]. The purpose is to generate  $H_2O_2$  *in-situ* from  $O_2$  reduction via Eq. (II.4) which is one of the reagents (with Fe<sup>2+</sup>) of Fenton process provided in Eq. (II.7).

$$Fe^{2+} + H_2O_2 \to Fe^{3+} + \bullet OH + OH^-$$
 (II.7)

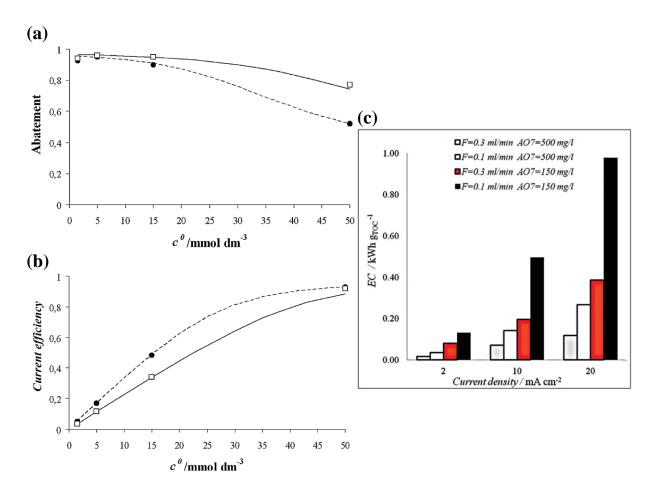
Furthermore, carbonaceous gas diffusion electrode has also been used to even improve H<sub>2</sub>O<sub>2</sub> production by maximizing the contact between cathode, O<sub>2</sub> (by supplying air flow) and electrolyte [15]. An anode possessing low overpotential for O<sub>2</sub> evolution is used for EF, e.g., Pt, RuO<sub>2</sub> and/or IrO<sub>2</sub>. These anodes are called active anodes, because •OH are strongly abided to the metal by chemisorption [17]. The metal M may form higher oxide MO and its decomposition releases O<sub>2</sub>. On an active anode, water oxidation produces dissolved O<sub>2</sub> which is the precursor of H<sub>2</sub>O<sub>2</sub> following Eq. (II.3). Thus, the EF reagents are electrocatalytically regenerated (Eqs. (II.4) and II.8) in the electrolytic cell.

$$Fe^{3+} + e^{-} \to Fe^{2+}$$
 (II.8)

A coupling of EAOP can be done in a single electrochemical cell simply by changing the electrode material. Successive attempts have been made in submillimetric reactor configuration for example during the degradation of chloroacetic acid [5] and Acid Orange 7 organic dye [6] by coupled EO-EF using BDD and compact graphite as anode and cathode respectively. A summary of different electrode materials used in EAOPs operated in microfluidic condition is given in **Table I.1**.

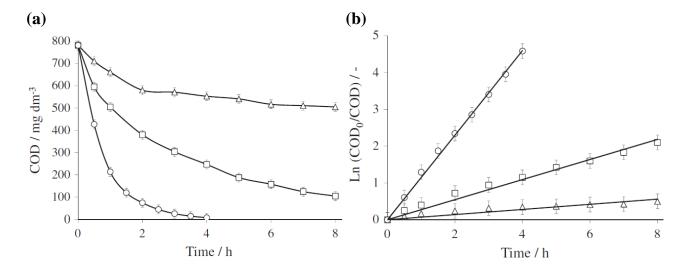
#### II.1.5. Influence of initial concentration of pollutant

The concentration of pollutant to be treated in the effluent can have an impact on the kinetics and efficiency of its degradation [4]. The results of EO of formic acid in a microfluidic reactor at varying formic acid concentrations, as plotted in Fig. II.4, could explain the mechanism [4]. In the lower range of concentrations ( $\leq 5$  mmol L<sup>-1</sup>), high pollutant oxidation was achieved and it was independent of applied current. The degradation process was supposed to be controlled by mass transfer. Contrastingly, using higher formic acid concentration ( $\geq 15$  mmol L<sup>-1</sup>), a decrease of the degradation yield was noted but  $\eta$  increased (Eq. (II.1)). It suggested that the availability of pollutant near the anode surface was no longer the limiting factor, meaning that the kinetics was therefore limited by the charge transfer. Elsewhere, the degradation efficiency of organic dye was evaluated by either EO or EF process in microfluidic configurations [13]. Once again, lower organic abatement at higher concentration was observed but with drastic decrease in energy consumption (Fig. II.4(c)). The latter was attributed to lower impact of parasitic oxidation reactions and therefore to the higher  $\eta$ .



**Fig. II.4.** Formic acid removal (a) and the associated  $\eta$  values (b) at varying initial formic acid concentrations (adapted from Scialdone et al. [4]). ( $\bullet$ ): 6.7 and ( $\square$ ): 10 mA cm<sup>-2</sup>. (c) Variation of *SPC* during organic dye (acid orange 7 (AO7)) degradation at different dye concentrations and  $j_{app}$  (adapted from Scialdone et al. [13]).

When high  $j_{app}$  was used to treat highly concentrated wastewater particularly using BDD as anode, the organic oxidation would be limited by mass transfer [16]. It is because sufficient electrical charge was supplied, which provided sufficient amount of  ${}^{\bullet}$ OH to quickly and efficiently degrade the organics. Figure II.5 gives an example of such case where landfill leachate waste was electro-oxidized using different anodes at 2 A [19]. Consequently, organic degradation followed and could be modeled using a pseudo first-order kinetic rate, when its oxidation was independent of initial organic content.



**Fig. II.5**. (a) COD abatement and (b) pseudo first-order kinetic plot during the EO of landfill leachate wastewater using ( $\triangle$ ): TiRuSnO<sub>2</sub>, ( $\square$ ): PbO<sub>2</sub> and ( $\bigcirc$ ): BDD anodes. *I*: 2 A. Adapted from Panizza et al. [19].

Table II.1. Literature review on the application of microfluidic reactors to treat wastewater.

Pollutant	EAOP	Mode	Electrode materials	$d_{ m elec}$ / $\mu { m m}$	Operating conditions	Ref
Oxalic acid	ЕО	Flow-by	Anode: BDD (9 cm <sup>2</sup> ) Cathode: Nickel (9 cm <sup>2</sup> )	50, 360 and 750	Oxalic acid: $360 - 900 \text{ mg L}^{-1}$ $j_{\text{app}}$ : $4.44 - 1.67 \text{ mA cm}^{-2}$ ( $40 - 150 \text{ mA}$ ) Flow rate: $0.5 \text{ mL min}^{-1}$ , $Na_2SO_4$ : $0 \text{ or } 5 \text{ mg}$ $L^{-1}$ , pH: $2$	[20]
Formic acid	ЕО	Flow-by	Anode: BDD (2.7 cm <sup>2</sup> ) Cathode: Nickel (2.7 cm <sup>2</sup> )	50 and 75	Formic acid: $69 - 2300 \text{ mg L}^{-1}$ $j_{app}$ : $2.2 - 40 \text{ mA cm}^{-2}$ ( $6 - 108 \text{ mA}$ ) Flow rate: $0.05 - 0.30 \text{ mL min}^{-1}$ , pH: 2	[4]
Acid Orange 7	EF	Flow-by	Anode: $Ti/IrO_2-Ta_2O_5$ (5 cm <sup>2</sup> ) Cathode: Compact graphite (5 cm <sup>2</sup> ) or Carbon felt (5 – 6 cm <sup>2</sup> )	7, 120 and 240	Acid orange 7: 150 mg L <sup>-1</sup> $j_{app}$ : 1 – 15 mA/cm <sup>2</sup> (5 – 75 mA) Flow rate: 0.1 mL min <sup>-1</sup> , Na <sub>2</sub> SO <sub>4</sub> : 5 g L <sup>-1</sup> , Fe <sup>2+</sup> : 76 mg L <sup>-1</sup> , volume: 50 mL	[3]
Chloroacetic acid	EO EF EO-EF	Flow-by	EO: Anode: BDD, Cathode: Stainless steel  EF: Anode: Ti/IrO <sub>2</sub> -Ta <sub>2</sub> O <sub>5</sub> , Cathode: Compact graphite  EO-EF: Anode: BDD, Cathode: Compact graphite  All electrode surface area: 5 cm <sup>2</sup>	50 and 100	Chloroacetic acid: 473 mg L <sup>-1</sup> $j_{app}$ : 1 – 20 mA cm <sup>-2</sup> (5 – 100 mA) Flow rate: 0.1 – 0.6 mL min <sup>-1</sup> , EF: 0 or 76 mg L <sup>-1</sup> Fe <sup>2+</sup> , pH: 3 (EF)	[5]
Acid Orange 7	EO EF Indirect oxidation by active chlorine (IOAC)	Flow-by	EO: Anode: BDD, Cathode: Nickel EF: Anode: Ti/IrO <sub>2</sub> -Ta <sub>2</sub> O <sub>5</sub> , Cathode: Compact graphite IOAC: Anode: Ti/IrO <sub>2</sub> -Ta <sub>2</sub> O <sub>5</sub> , Cathode: Nickel All electrode surface area: 5 cm <sup>2</sup>	50, 75, 120 and 240	Acid Orange 7: 150 mg L <sup>-1</sup> $j_{app}$ : 2 – 14 mA cm <sup>-2</sup> (10 – 70 mA) Flow rate: 0.1 – 0.4 mL min <sup>-1</sup> , EF: 34 – 152 mg L <sup>-1</sup> Fe <sup>2+</sup> , pH: 7 (EO, IOAC) and 3 (EF), NaCl: 993 mg L <sup>-1</sup> (IOAC)	[6]

Chapter II

Pollutant	EAOP	Mode	Electrode materials	$d_{ m elec}$ / $\mu { m m}$	Operating conditions	Ref
Tetrachloroethane	EO ER EO-ER	Flow-by	EO: Anode: BDD, Cathode: Nickel ER: Anode: DSA, Cathode: Silver EO-ER: Anode: BDD, Cathode: Silver All electrode surface area: 3-4 cm <sup>2</sup>	50 and 75	Tetrachloroethane: $151 - 285 \text{ mg L}^{-1}$ $j_{\text{app}}$ : $3 - 15 \text{ mA cm}^{-2} (9 - 45 \text{ mA})$ Flow rate: $0.1 - 0.4 \text{ mL min}^{-1}$ , pH: 2	[14]
Conversion of dichloroacetic acid to chloroacetic acid	ER	Flow-by	Anode: Ti/IrO <sub>2</sub> -Ta <sub>2</sub> O <sub>5</sub> , Cathode: Compact graphite  Configuration in series: Surface area: 4 cm <sup>2</sup> (each electrode)  Configuration in stack: Surface area: 6 cm <sup>2</sup> (each chamber)	100	Dichloroacetic acid: $13 - 64.5 \text{ mg L}^{-1}$ $j_{\text{app}}$ : $10 - 48 \text{ mA cm}^{-2}$ Flow rate: $0.05$ , $0.1$ and $0.2 \text{ mL min}^{-1}$	[21]
	ЕО		EO: Anode: BDD, Cathode: Nickel		Acid Orange 7: 150 and 500 mg L <sup>-1</sup> <b>EO</b> : $j_{app}$ : 2 – 20 mA cm <sup>-2</sup>	
Acid Orange 7	EF	Flow-by	<b>EF</b> : Anode: DSA, Cathode: Graphite Configuration: single, 3 in series or in series with between 2 EAOPs	EO: 50 EF: 120	Flow rate: $0.1 - 0.3 \text{ mL min}^{-1}$ <b>EF</b> : $j_{app}$ : $1 - 20 \text{ mA cm}^{-2}$	[13]
	EF-EO		All electrode surface area: 4 cm <sup>2</sup>	L1. 120	Flow rate: 0.1 mL min <sup>-1</sup> , Fe <sup>2+</sup> : 76 mg L <sup>-1</sup> , pH: 3	
Total organic carbon in reclaimed industrial WWTP effluent	ЕО	Flow-by	Anode: BDD (3.75 cm <sup>2</sup> ) Cathode: Nickel (3.75 cm <sup>2</sup> )	50	TOC <sub>0</sub> : 210 mg L <sup>-1</sup> $j_{app}$ : 5.3 – 53.3 mA cm <sup>-2</sup> (20 – 200 mA) Flow rate: 0.1 – 0.5 mL min <sup>-1</sup> , pH: 6.2, volume: 50 mL	[7]
Diuron herbicide	ЕО	Flow-by	Anode: Graphite sheet (3 cm <sup>2</sup> ) Cathode: Stainless steel (3 cm <sup>2</sup> )	250 – 750	Diuron: $10 \text{ mg L}^{-1}$ $j_{app}$ : $0.16 - 0.64 \text{ mA cm}^{-2} (0.5 - 2 \text{ mA})$ pH: $3-10$ , conductivity: $6.7 - 1000 \mu\text{S cm}^{-1}$	[9]
Paracetamol in synthetic solution and reclaimed municipal WWTP effluent	ЕО	Flow-by	Anode: BDD (50 cm <sup>2</sup> ) Cathode: Carbon felt (50 cm <sup>2</sup> )	50 – 1000	Paracetamol: 15 mg L <sup>-1</sup> $j_{app}$ : 2 to 12 mA cm <sup>-2</sup> pH: 3 or neutral, Na <sub>2</sub> SO <sub>4</sub> : 0.14, 0.57 and 1.42 g L <sup>-1</sup> , conductivity: 230 to 2000 $\mu$ S cm <sup>-1</sup> , flow rate: 0.43 L min <sup>-1</sup> , volume: 200 or 500 mL	[11]

Chapter II

Pollutant	EAOP	Mode	Electrode materials	$d_{ m elec}$ / $\mu { m m}$	Operating conditions	Ref
Paracetamol in synthetic solution	ЕО	Flow- through	Anode: Perforated BDD (14 cm <sup>2</sup> ) Cathode: Carbon felt (14 cm <sup>2</sup> )	500	Paracetamol: 15 mg L <sup>-1</sup> $j_{app}$ : 4 mA cm <sup>-2</sup> Na <sub>2</sub> SO <sub>4</sub> : 570 mg L <sup>-1</sup> (850 $\mu$ S cm <sup>-1</sup> ), pH: neutral, volume: 4 L	[12]
Clopyralid in simulated soil washing effluent	ЕО	Flow- through	Anode: BDD mesh (33 cm <sup>2</sup> ) Cathode: Perforated plate stainless steel (33 cm <sup>2</sup> )	400	Clopyralid: $100 \text{ mg L}^{-1}$ $j_{\text{app}}$ : $10 \text{ mA cm}^{-2}$ Flow rate: $1.67 \text{ L min}^{-1}$ , volume: $1 \text{ L}$	[22]
Clopyralid in simulated soil washing effluent	EO	Flow- through	Anode: BDD (50 cm <sup>2</sup> ) or Ti/RuO <sub>2</sub> -IrO <sub>2</sub> MMO mesh (53 cm <sup>2</sup> )  Cathode: Perforated stainless steel plate  (33 cm <sup>2</sup> )	400	Clopyralid: $100 \text{ mg L}^{-1}$ $j_{app}$ : $10 \text{ and } 100 \text{ mA cm}^{-2}$ Flow rate: $1.67 \text{ L min}^{-1}$ , volume: $1 \text{ L}$	[8]
Clopyralid in simulated soil washing effluent	EF	Flow- through	Anode: Ti/RuO <sub>2</sub> -IrO <sub>2</sub> MMO (53 cm <sup>2</sup> ) or BDD mesh (50 cm <sup>2</sup> ) Cathode: CB/PTFE-RVC or CB/PTFE- Al (33 cm <sup>2</sup> )	400	Clopyralid: $100 \text{ mg L}^{-1}$ $j_{app}$ : 20, 30 and 50 mA cm <sup>-3</sup> Flow rate: $1.17 \text{ L min}^{-1}$ , Fe <sup>2+</sup> : 28 and 112 mg L <sup>-1</sup> , Na <sub>2</sub> SO <sub>4</sub> : 1 or 7.1 g L <sup>-1</sup> , pH: 3, volume: 750 mL	[23]
Clopyralid in simulated soil washing effluent	EF	Flow- through	Anode: Ti/RuO <sub>2</sub> -IrO <sub>2</sub> MMO (33 cm <sup>2</sup> ) Cathode: CB/PTFE-Al foam (33 cm <sup>2</sup> ), Al foam thickness: 5 mm. For scale-up study: 10 and 15 mm	150	Clopyralid: $100 \text{ mg L}^{-1}$ $j_{\text{app}}$ : $10 - 60 \text{ mA cm}^{-3}$ Flow rate: $2.67 \text{ L min}^{-1}$ , $Na_2SO_4$ : $7.1 \text{ g L}^{-1}$ , volume: $2.25 \text{ L}$ , pressurized air: $6 \text{ bars}$	[24]
Pretreated soil washing wastewater containing clopyralid herbicide	EO EF	Flow- through	EO: Anode: BDD mesh, Cathode: perforated stainless steel plate EF: Anode: BDD mesh, Cathode: CB/PTFE-Al foam All electrode surface area: 33 cm <sup>2</sup>	150	400 g of soil polluted with 100 g kg <sup>-1</sup> of clopyralid washed with 1 L water $j_{app}$ : 10 – 100 mA cm <sup>-2</sup> Flow rate: 0.4 – 1.6 L min <sup>-1</sup> , EF: 28 mg L <sup>-1</sup> Fe <sup>2+</sup> , pH: 3 (EF)	[10]

<u>Abbreviations</u>: EO: Electro-oxidation, ER: electro-reduction, EF: electro-Fenton, BDD: boron-doped diamond, DSA: dimensionally stable anode, CB: carbon black, PTFE: polytetrafluoroethylene, RVC: reticulated vitreous carbon, Al: aluminum,  $d_{\text{elec}}$ : interelectrode distance,  $j_{\text{app}}$ : applied current density, TOC: total organic carbon.

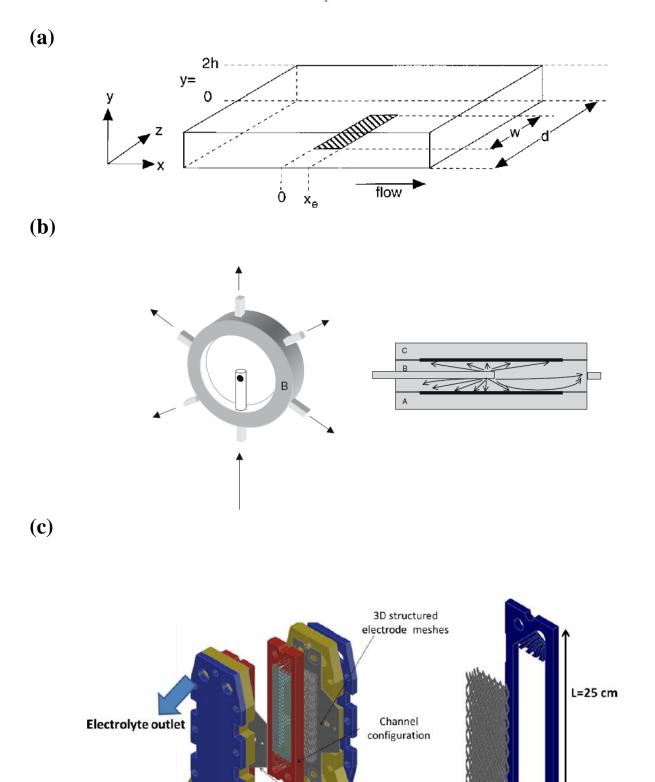
#### II.1.6. Design and modularity of microfluidic electrochemical cell

#### II.1.6.1. Mass transfer in submillimetric electrochemical reactors

The application of submillimetric reactor configurations requires further in-depth understanding regarding the mass transfer phenomenon at the core of the electrochemical cell. Thin film reactors [25-27] introduced in the early 80s consisted of a plate electrode (i.e., the working electrode) most of the time positioned horizontally and were equipped with a counter electrode at the downstream of the reactor (Fig. II.6(a)). The reference electrode RE was positioned close to the working electrode, upstream if a 3-electrode configuration was adopted [27]. This thin film setup is not appropriate to receive wastewater effluent. Flow-by filter-press or flow-through reactor design is more apt for wastewater application at pilot or even industrial scale (e.g., Figs. II.6(b)-(c)).

Correlation in the form of  $Sh = aRe^bSc^cLe^d$  assuming Lévêque approximation is common in parallel-plate electrochemical reactors in the literature. It allows a dimensionless expression of the mass transfer coefficient  $(k_m)$  with dimensionless fluid properties as well as dimensionless reactor configurations under respective hydrodynamic regime of liquid flow. As a result, the extrapolation of  $k_m$  for a given reactor dimension during a scale up or scale down is possible. **Table II.2** regroups the coefficients of a, b, c and d of Sh correlation in parallel-plate reactor design under laminar regime. Some of the cited works used turbulence promoter to enhance mass transfer but the flow regime remained laminar [28-31]. **Table II.3** groups the values of  $k_m$  determined in parallel-plate electrochemical cells. Surprisingly, there is lack of data with regard to the  $k_m$  values under submillimetric electrochemical design. To the best of authors' knowledge, only Scialdone et al. [4] and Pérez et al. [8] evaluated the  $k_m$  value in their respective submillimetric reactor setup.

Hence, the establishment of mass transfer correlations under submillimetric condition would be beneficial to the body of knowledge since it concerns not only the application in environmental field but also in other electrochemistry areas. It would allow a convenient estimation of  $k_{\rm m}$  for a scale up or scale down of an electrochemical reactor operated in comparable hydrodynamic regime. Since the correlation is expressed in dimensionless form, the correlation remains valid to other microfluidic reactor geometries of interest.



**Fig. II.6**. (a) Thin-film channel flow cell from reproduced from Cooper et al. [27] and more recent parallel-plate electrochemical reactors such as (b) parallel-plate with impinging inlet reproduced from Martínez-Huitle et al. [32] and (c) commercial parallel-plate filter-press type reactor reproduced from Cruz-Díaz et al. [30].

Electrolyte inlet

Counterelectrode

S=1.1 cm

**Table II.2**. Coefficients for *Sh* correlation in millimetric parallel-plate electrochemical reactors within the given range of *Re* and *Sc* numbers.

C 40					Re validity	Sc validity	ъ.
Configuration	а	b	c	d	ranges	ranges	Ref.
Parallel plate without turbulence promoter	0.18	0.73	0.33	-	500 - 2200	1494	[28]
Parallel plate with turbulence promoter	0.71	0.55	0.33	-	500 - 2200	1494	[28]
Parallel-plate reactor	2.54	0.33	0.33	-	< 2300	-	[33]
Parallel-plate reactor	2.54	0.33	0.29	0.33	200 - 2000	960 - 3300	[34, 35]
Parallel-plate reactor without turbulence promoter	0.28	0.70	0.33	-	148 - 6109	1588	[36]
Flow cell Diacell®	0.69	0.36	0.33	-	25 - 100	1780	[37]
Flow cell Diacell®	0.14	0.70	0.33	-	100 - 2500	1780	[37]
Parallel-plate reactor	0.56 – 0.71	0.36 – 0.37	0.33	-	125 - 3550	1469	[38]
Parallel-plate reactor with turbulence promoter	2.77	0.41	0.41	0.41	839 - 1525	1420	[29]
Parallel-plate reactor with turbulence promoter	0.076	0.25	0.33	-	90 - 180	3774	[30]
Parallel-plate reactor with turbulence promoter	0.14 – 0.27	0.67	0.33	-	55 - 2100	1633	[31]
Parallel-plate reactor without turbulence promoter	0.19	0.81	0.33	-	1250 - 6900	1633	[39]

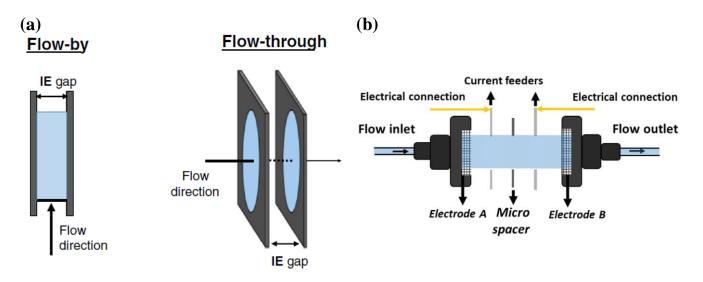
**Table II.3**. Reactor designs and their evaluated range of  $k_{\rm m}$  values.

Type of reactor	Cell configuration	$k_{\rm m}$ / m s <sup>-1</sup>	Ref.
Parallel-plate / flow-by	d <sub>elec</sub> : 32 mm Geometry: Rectangular Area: 100 cm <sup>2</sup>	$0.86 - 3.11 \times 10^{-5}$	[36]
Parallel-plate / flow-by	Flow rate: 4 to 24.7 L min <sup>-1</sup> $d_{\text{elec}}: 11 \text{ mm}$ Geometry: Rectangular Area: 247.5 cm <sup>2</sup> Flow rate: $0.05 - 0.6 \text{ L min}^{-1}$	$1.44 - 1.75 \times 10^{-6}$	[30]
Parallel-plate / flow-by	$d_{\text{elec}}$ : 10 mm  Geometry: Circular  Area: 63.5 cm <sup>2</sup> Flow rate: 2.75 L min <sup>-1</sup>	2.10 × 10 <sup>-5</sup>	[40]
Parallel-plate with impinging inlet	d <sub>elec</sub> : 10 mm Geometry: Circular Area: 63.5 cm <sup>2</sup> Flow rate: 1.10 L min <sup>-1</sup>	2.97 × 10 <sup>-5</sup>	[32]
Parallel-plate / flow-by	$d_{\rm elec}$ : 10 mm Geometry: Circular Area: 63.5 cm <sup>2</sup> Flow rate: 9.1 L min <sup>-1</sup>	$2.00 \times 10^{-5}$	[41]
Parallel-plate / flow-by	$d_{ m elec}$ : 9 mm Geometry: Circular Area: 78 cm <sup>2</sup> Flow rate: 1.25 L min <sup>-1</sup>	$2.83 \times 10^{-5}$	[42]
Parallel-plate / flow-by	$d_{\rm elec}$ : 7 mm Geometry: Circular Area: 63.2 cm <sup>2</sup> Flow rate: 1.4 - 135 L min <sup>-1</sup>	$0.23 - 3.3 \times 10^{-5}$	[37]
Parallel-plate / flow-by	$d_{\rm elec}$ : 5.5 mm Geometry: Porous or non-porous rectangular Volumetric area: 30, 58 and 517 cm <sup>-1</sup> Flow rate: 1.52 – 3 L min <sup>-1</sup>	$5.42 - 579 \times 10^{-8}$	[43]
Parallel-plate / flow-by	$d_{\rm elec}$ : 5 mm Geometry: Rectangular Area: 80 cm <sup>2</sup> Flow rate: 27 to 84 L min <sup>-1</sup>	$0.30 - 1.20 \times 10^{-5}$	[44]
Parallel-plate / flow-by	$d_{\rm elec}$ : 1 mm Geometry: Circular Area: 70 cm <sup>2</sup> Flow rate: 5.6 to 18.3 L min <sup>-1</sup>	1.60 × 10 <sup>-5</sup>	[45]
Parallel-plate / flow-by	$d_{\rm elec}$ : $0.05 - 0.075$ mm Geometry: Rectangular Area: $2.7~{\rm cm}^2$ Flow rate: $0.05$ to $0.3 \times 10^{-3}$ L min <sup>-1</sup>	$*2.15 - 3.22 \times 10^{-5}$	[4]
Parallel-plate / flow-through	d <sub>elec</sub> : 0.4 mm Geometry: Porous, circular Area: 52.8 cm <sup>2</sup> Flow rate: 0.5 to 1.5 L min <sup>-1</sup>	1.45 × 10 <sup>-5</sup>	[8]

<sup>\*</sup>estimated from *Sh* number used in their work

# II.1.6.2. Microfluidic electrochemical reactor in wastewater treatment application

To date, two electrochemical reactor designs have been adopted to treat wastewater in microfluidic conditions as summarized in **Table II.1**. The first one is a flow-by design and the other one is a flow-through configuration as illustrated in **Fig. II.7**.



**Fig. II.7**. (a) Difference between flow-by and flow-through electrolytic flow and (b) scheme of flow-through microfluidic cell adopted by Pérez et al. [22].

In the flow-by configuration, the electrolyte flowed in between two planar electrodes separated by a micrometric distance [22]. In the flow-through mode, the electrodes were porous, typically meshes, which allowed the electrolyte to pass through them [22]. The two meshes of anode and cathode were separated by a spacer, micrometric in thickness, which gave microfluidic characteristics to the reactor [22].

A comparison in terms of pollutant degradation as well as economic performance between the two designs has been performed and it has been illustrated in **Fig. II.8**. It can be observed that both degradation of clopyralid and the mineralization of Total Organic Carbon (TOC) were faster using flow-through mode at both applied current investigated (**Figs. II.8(a)-II.8(b)**). At 10 mA cm<sup>-2</sup>, complete removal of clopyralid and TOC were obtained after 10.1 and 11.4 Ah dm<sup>-3</sup> respectively using flow-by mode, whilst they required 2.4 and 2.7 Ah dm<sup>-3</sup> in flow-through mode. Lower *SPC* was also obtained in the flow-through configuration, which was evaluated at 12 kWh m<sup>-3</sup> using 10 mA cm<sup>-2</sup> against 61 kWh m<sup>-3</sup> in flow-by mode working at similar *j*<sub>app</sub>. Better energetic performance was attributed not only to lower cell voltage but also to improved mass transfer [**8, 22**]. With 3-dimensional mesh electrodes in flow-through cell, higher geometric area and specific surface of

electrode could be achieved over planar electrodes and more local turbulence occurred in the mesh. Therefore, the reactor design was more compact and  $k_{\rm m}$  was higher compared to the flow-by design.

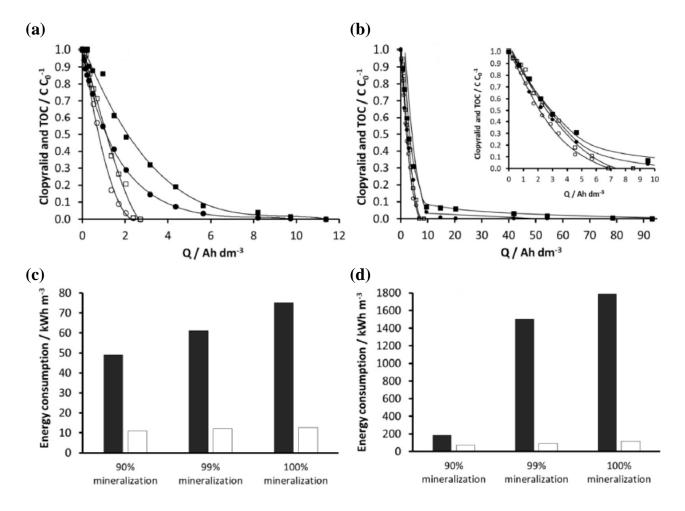


Fig. II.8. Comparison of clopyralid pesticide ( $\bullet$ ,  $\bigcirc$ ) and TOC ( $\blacksquare$ ,  $\square$ ) removal using flow-by ( $\bullet$ ,  $\blacksquare$ ) and flow-through ( $\bigcirc$ ,  $\square$ ) microfluidic reactors by applying 10 ((a), (c)) and 100 ((b), (d)) mA cm<sup>-2</sup>. Power consumption of flow-by (black bar) and flow-through (white bar) during the process to reach different mineralization degrees at 10 mA cm<sup>-2</sup> (c) and 100 mA cm<sup>-2</sup> (d). Flow-by  $d_{\text{elec}}$ : 3000  $\mu$ m and flow-through  $d_{\text{elec}}$ : 400  $\mu$ m. Anode: BDD and cathode: stainless steel. Adapted from Pérez et al. [8].

The fact that the electrochemical microreactor implies small volume within the cell, the productivity of the process using such reactor design is often questionable. To address this problem, several authors have tempted operating the reactors in series. For example, up to three flow-by microfluidic reactors were operated in series (**Fig. II.9(a**)) to degrade an organic dye up to 500 mg L<sup>-1</sup> in concentration [13]. The results are presented in **Fig. II.10**. It was shown that color and organics removal as well as treatment productivity could effectively be improved by operating the cells in series as expected. A coupling of two microfluidic reactors combining EF and EO were as well modulated. The first process of EF was aimed to reduce organic content using cheap electrodes while using low energy consumption, whilst from the second EO process, the strong oxidation by **OH** was intended to achieve total organics degradation. Optimal coupling was obtained by adapting different

applied parameters required by each operation [13]. An assay has also been made to increase  $H_2O_2$  production in a flow-through microfluidic cell [24]. It was done by using thicker cathode materials (5, 10 and 15 mm in length, which gave 16.5 to 49.5 cm<sup>3</sup> in volume (depicted in Fig. II.9(b))) hence sizing up the geometrical volume of electrochemical reactor. The authors reported having a proportional increase in  $H_2O_2$  production rate with the increasing cathode thickness. However, ohmic resistance was reported to increase slightly when the thickness of cathode was increased. Cell voltage also increased with electrode thickness. As a result, higher cell voltage entailed an increase in SPC.

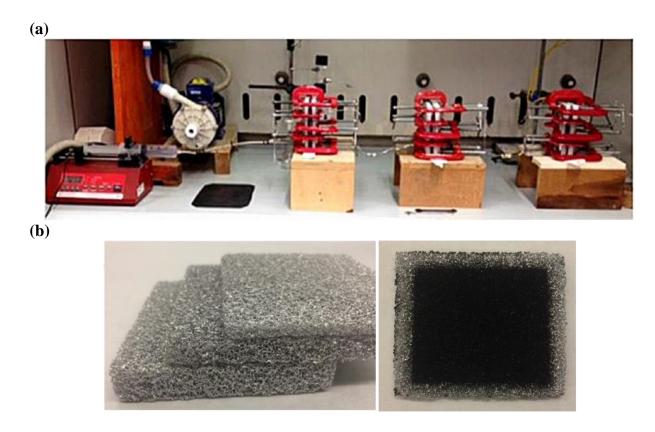
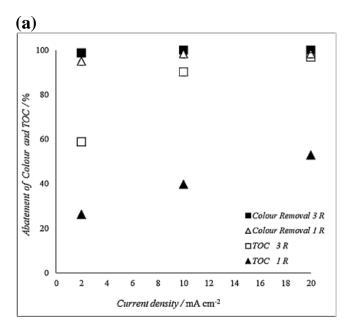
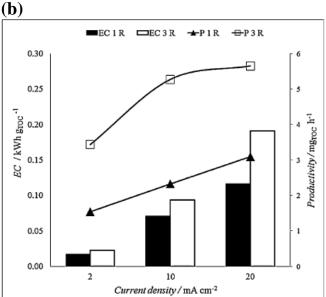


Fig. II.9. (a) Flow-by microfluidic reactors operating in series (adapted from Sabatino et al [13]) and (b) cathode material thicknesses used in flow-through microfluidic cell from Pérez et al [24].



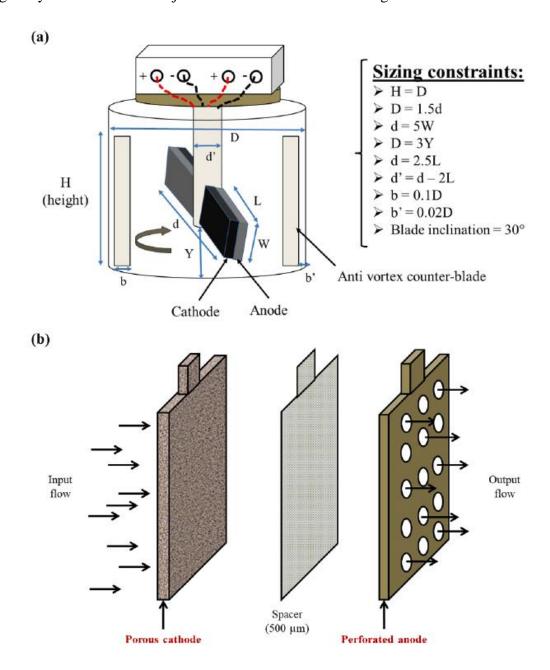


**Fig. II.10**. (a) Color and TOC removal as well as (b) energy consumption and productivity for flow-by microfluidic reactor working in single or in series at different  $j_{app}$ . Adapted from Sabatino et al. [13].

Recently, another innovative microreactor design has been introduced, which could be particularly applicable in the field of wastewater treatment [12]. The reactor is illustrated in Fig. II.11. This design was proposed to reduce the mass transfer limitation often encountered in conventional macroreactors as well as to address the low treatment capacity and the clogging issue related to microfluidic reactors. It has been called reactive electro-mixing reactor and it consists of impellers, which themselves are composed of thin film electrochemical reactors. The reactors were set on motion with the rotation of the reactive electro-blades. One reactor can accommodate several impellers, which thus means several microreactors in one macroreactor. In the meanwhile, the macroreactors could be operated in series to significantly increase the productivity.

In summary, only a handful of publications exist in the literature regarding the use of microfluidic reactors in the field of water and wastewater treatments. Low productivity and clogging issues remain a big hurdle for the application of submillimetric reactors in large scale. These challenges are more present when real wastewater is to be treated. In the scope of reclaimed municipal or industrial wastewater treatments for water reuse, the effluent often contains colloidal and suspended solids that would need a preliminary separation step. Moreover, the organic content in the effluent is low. When the degradation of micropollutant is intended, these low concentrations could quickly reduce the process efficacy. This is where the applicability of electrochemical microreactors in the field of environmental engineering would be of interest. It could lift the mass transfer constraint encountered in conventional reactors while avoiding the use of external chemical reagents. Thus, the investigation of different behaviors between millimetric and microfluidic configuration is the grand theme of the

thesis. The occurrence of electro-precipitation, particularly in microfluidic features, has never been investigated yet. It will be the subject of discussion in the incoming sections.



**Fig. II.11**. (a) Schematic diagram of reactive electro-mixing reactor and (b) an reactive electro-blade acting as flow-through microreactor. Reproduced from Mousset [12].

#### II.2. Concomitant cathodic electro-precipitation during an electrolytic process

Calcareous scaling deposition that takes place naturally due to  $Ca^{2+}$  and  $Mg^{2+}$  content of natural water (water hardness) is a permanent problem. Technical and economic issues pertaining calcareous scaling in industrial plants and domestic equipment have triggered researches on this topic in early ages, including via electrochemical methods which has also been called accelerated scaling process [46-50]. This method implies the reduction of dissolved  $O_2$  (Eq. (II.9)) and water (Eq. (II.10)) to favor the mineral deposition on cathode of an electrochemical cell [51, 52].

$$O_2 + 2H_2O + 4e^- \rightarrow 4OH^-$$
 (II.9)

$$2H_2O + 2e^- \rightarrow H_2 + 2OH^-$$
 (II.10)

Ever since, the cathodic electrochemical precipitation has been widely documented in the literature and it concerns three major domains of application namely cathodic protection (Table II.4), phosphorus recovery (Table II.5) and water softening (Table II.6). On one hand, the deposition on cathode is deliberate to achieve the intended goal. On the other hand, such as in the case with EAOPs, the cathodic deposition is a major drawback, which would decrease the process efficiency during a long operation. Mineral scaling could result in economic challenges in all kinds of electrochemical reactors and the main influencing parameters are exposed hereinafter including the submillimetric features that will be adopted in this thesis.

## II.2.1. Applied cathode potential/current density

# II.2.1.1. Relation between cathode potential, local alkalization and mineral scaling

Corrosion of a metal occurs when the metal (M) spontaneously releases its ion (M<sup>Z+</sup>) into the solution that it is in contact with. It is a spontaneous reaction when two metal elements are present, with one having more tendency to release electrons (less noble metal) acting as anode. Corroded metal loses its tensile strength, becomes brittle and has short life span. That is the reason why several authors have investigated the corrosion behavior of soft metal and carbon metal that have widely been used in marine infrastructure since they present more economic value. For instance, the formation of corrosion product by applying a corrosion potential has been investigated by Rakitin et al. [53] and Gabrielli et al. [46] on ferrous metal and also by Ben Amor et al. [54], Bousselmi et al. [55] and Marin-Cruz et al. [56] on carbon steel. Furthermore, by applying a more negative potential that corresponds to the potential of the reduction of dissolved O<sub>2</sub> or water (Eqs. (II.9)-(II.10)), the reduction reactions producing OH<sup>-</sup> are accelerated. This continuous production of OH<sup>-</sup> induces high

local pH increase at the cathode vicinity. In the presence of carbonates, Ca<sup>2+</sup> and Mg<sup>2+</sup> in solution, mineral deposits are therefore formed on the cathode surface (**Eqs. (II.11)-(II.13))** [50, 57-59].

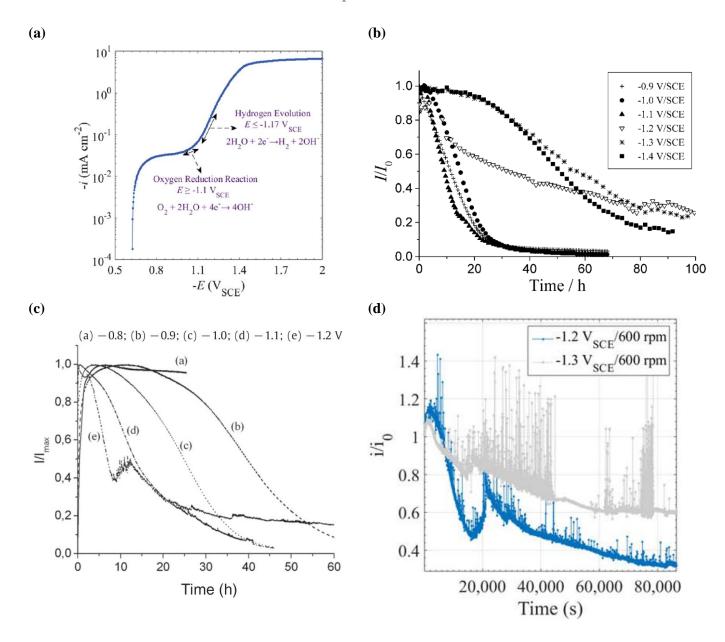
$$Mg^{2+} + 2OH^- \leftrightarrow Mg(OH)_2 \tag{II.11}$$

$$OH^- + HCO_3^- \leftrightarrow H_2O + CO_3^{2-}$$
 (II.12)

$$Ca^{2+} + CO_3^{2-} \leftrightarrow CaCO_3 \tag{II.13}$$

These precipitates, also referred as scaling, serve as the basis for cathodic protection application, which has vastly been applied in underwater pipelines, marine transportation and above/under water infrastructures. It offers an excellent barrier towards further degradation of metal in contact with water. It significantly reduces the diffusion of dissolved O<sub>2</sub> towards the metal surface hence hindering further corrosion.

It has been shown across literature that the kinetics, quantity and crystallography of calcareous deposit are heavily dependent on the applied potential [57, 59-61]. Figure II.12 plots normalized chronoamperometric curves during the formation of calcareous scaling at different applied potentials. According to Figs. II.12(b)-II.12(c) a potential of -1.2 V vs. saturated calomel electrode (SCE) represented a potential threshold [59, 60]. From -0.9 to -1.1 V/SCE, the applied potentials were on the plateau of the O<sub>2</sub> reduction producing OH<sup>+</sup> (Eq. (II.9)) according to the voltammogram plotted in Fig. II.12(a) [62]. The lower the potential, the higher the rate of OH<sup>+</sup> production, hence faster scaling (Eq. (II.13)) occurred. As a result, cathode surface was completely covered by the passivated film, thus the residual current at the end of the electrolysis was almost nil. At -1.2 V/SCE and below, the applied potential reached the region of water reduction (Eq. (II.10)) [62]. OH<sup>+</sup> production occurred with concurrent HER. The evolved H<sub>2</sub> on the cathode surface induced partial detachment of calcareous scaling [60, 61]. It also led to a porous morphology of the scaling, which made the cathode surface not totally covered by the deposition. Consequently, higher residual current was noticed at the end of electrolysis (Figs. II.12(b)-II.12(d)).



**Fig. II.12**. (a) Voltammogram on steel in synthetic seawater from [62] and chronoamperometric curves during the formation of calcareous scaling at different applied potential, adapted from (b) Barchiche et al. [59], (c) Karoui et al. [60] and (d) Hoseinieh et al. [61].

### II.2.1.2. Gas evolution activity on electrode surface

When the applied potentials on cathode and anode surpass the potential of water reduction and oxidation respectively, HER (Eq. (II.10)) and O<sub>2</sub> evolution reaction (OER) ((Eq. (II.3)) occur. In the lower region of applied current, i.e., under the condition where HER and OER are already reached, the majority of gases are in dissolved form [63]. On a gas evolving electrode, it has been reported that there are periodic nucleation, growth and detachment of bubbles from electrode surface [63-65]. These periodic evolving and departing gas bubbles have been documented to enhance mass transfer locally on electrode surface [63, 64, 66-68]. The enhancement takes place firstly via microconvection owing to the evolving and departing gas bubbles from the electrode surface (Fig. II.13). Secondly,

the displacement of bubbles over the electrode surface induces a mechanism known as forced convection or macroconvection [63-65, 67, 68]. As the applied current density increases, the production of gases becomes more intense. The supersaturation of dissolved gas in the vicinity of electrodes increases too. The adhering gas bubbles could coalesce between each other to produce a film of gas bubbles, so-called gas curtain on electrode surface [66]. It can result in a significant increase in ohmic drop under certain configurations especially when the electrodes are brought close to each other [69-73]. The coalescence of dispersed gas bubbles might contribute towards gas film formation [74]. In such cases, slug flow (no longer dispersion) of bubbles, whose thicknesses were below 5 mm, was reported for example in a duct, [69]. The appearance of slug flow of gas bubbles was fact-driven and even more would be expected under the microfluidic setup [75].

The characterization and mechanistic understanding on electro-precipitation taking place inside submillimetric up to millimetric cells lack studies taking into account the gas evolution reactions in the precipitates stability. This will therefore be one of the objectives of the PhD thesis.

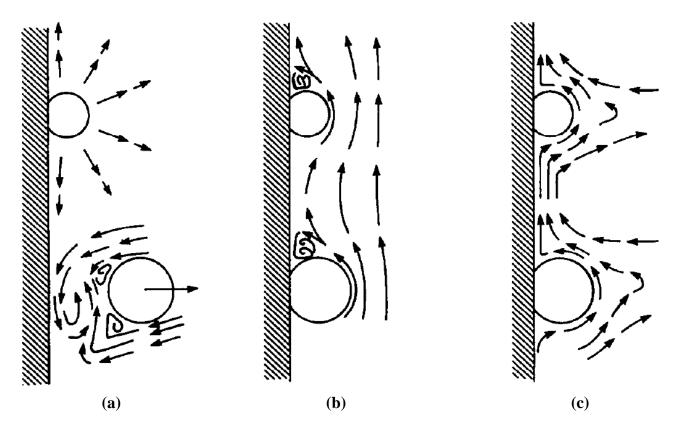
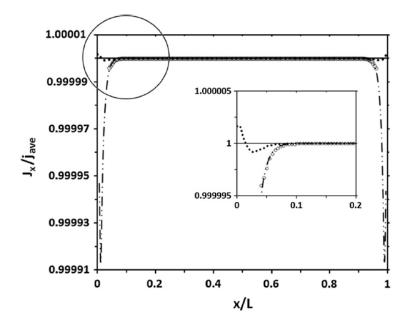


Fig. II.13. Bubbles flow pattern inducing mass transfer enhancement through (a) microconvection, (b) macroconvection and (c) free convection. Reproduced from Vogt et al. [66].

#### II.2.1.3. Current distribution on gas evolving electrode

The current distribution on electrode surface could be affected by the evolving gas. However, it has been reported a homogeneity of the current density distribution on large ( $64 \text{ cm}^2$ ;  $16 \times 14 \text{ cm} \times \text{cm}$ ) flat anode surface inside a commercial filter-press reactor with a pair of BDD and platinized titanium

electrodes with  $d_{\text{elec}}$  of 5 mm [76]. The result is depicted in Fig. II.14 in which it is affirmed a negligible drop of  $10^{-5}$  relative current density very closely to the border, where the electrolyte was in contact with anode-cathode separator.



**Fig. II.14.** Simulation of current density distribution during a typical EO operation producing \*OH on BDD anode during water oxidation in 1 M H<sub>2</sub>SO<sub>4</sub>. Potential sweep between 0.66 to 3 V/SHE to identify the region of \*OH production from H<sub>2</sub>O oxidation. Reproduced from Pérez et al. [76].

Then, microfluidic reactors might behave differently compared to millimetic reactors due to changes in interelectrode gap and hydrodynamic behavior. Thus, an investigation on the homogeneity of current distribution on electrode surface as well as the impact that it could have towards electrochemical reaction selectivity was performed [75]. The study was conducted in a microfluidic setup with the involvement of gas evolution in between electrodes. It was reported that the formation of bubbles inside the micrometric interelectrode space can perturb the spread of current density line across electrode surface unlike in the case with larger interelectrode gap. However, as long as the electrolyte flow was reasonably high (i.e., above 2.5 mL min<sup>-1</sup>) and that the bubble size was less than 50% of the  $d_{\rm elec}$  for fast-kinetic electrochemical reactions, the disturbance on the homogeneity of current density distribution was minimal. Thus, the selectivity of the desired reaction on electrode surface was ensured. Moreover, an average diameter of H<sub>2</sub> bubbles that evolved on electrode surface under the polarization of less than 10 mA cm<sup>-2</sup> was 26 µm [74]. The average diameter of gas bubble varied as function of  $j_{app}$  with eventual possibility of gas bubbles coalescence at relatively higher  $j_{app}$ (> 30 mA cm<sup>-2</sup>). Therefore, in function of  $j_{app}$ , adopted  $d_{elec}$  and electrolytic flow rate, the current distribution homogeneity and reaction selectivity could be impacted by the evolving gas. These features would then be looked upon during the selection of operating parameters.

Table II.4. Literature review on CaCO<sub>3</sub> and Mg(OH)<sub>2</sub> electro-precipitates on cathode surface.

Application	Precipitation	Electrochemical system	Electrolyte	Potential	Allotropy	Parameter	Ref
Calcareous scaling	CaCO <sub>3</sub>	WE: Fe disk (2 cm <sup>2</sup> ) CE: Pt RE: Ag/AgCl Volume: 80 mL	CaCl <sub>2</sub> or Ca(NO <sub>3</sub> ) <sub>2</sub> : 0.05 M NaCl, NaI, NaBr or NaNO <sub>3</sub> : 1 M NaHCO <sub>3</sub> : 2.1 g L <sup>-1</sup> Volume: 80 mL	Open circuit (corrosion potential)	Aragonite	• Anions (Cl <sup>-</sup> , I <sup>-</sup> , Br <sup>-</sup> and NO <sub>3</sub> <sup>-</sup> ) • Bicarbonates	[53]
Calcareous scaling	CaCO <sub>3</sub>	WE: Fe disk CE: unspecified RE: Saturated sulfate electrode (SSE)	Carbonically pure water, Ca <sup>2+</sup> : 200 mg L <sup>-1</sup> ,  Mineral drinking water,  Ca <sup>2+</sup> : 468 mg L <sup>-1</sup> Scaling inhibitor: aminotris(methylene-phosphonic)  (ATMP): 1 mg L <sup>-1</sup>	Open circuit (corrosion potential)	Aragonite	<ul> <li>Ca<sup>2+</sup> concentration</li> <li>Scaling inhibitor</li> </ul>	[46]
Calcareous scaling	CaCO <sub>3</sub>	WE: Pt disk (0.2 cm²) CE: unspecified RE: Saturated sulfate electrode (SSE)	Carbonically pure water, Ca <sup>2+</sup> : 20, 120 and 160 mg L <sup>-1</sup> KCl: 10 <sup>-4</sup> to 10 <sup>-2</sup> M K <sub>2</sub> SO <sub>4</sub> : 1 mM (blank)	Open circuit (corrosion potential) -1 and -1.5 V/SSE	Calcite	<ul><li> Ionic strength</li><li> Rotational speed</li></ul>	[47]
Calcareous scaling	CaCO <sub>3</sub>	WE: Gold disk (0.2 cm²)  CE: Pt  RE: Saturated sulfate electrode (SSE)	Carbonically pure water, Ca <sup>2+</sup> : 40, 80, 120 and mg L <sup>-1</sup> NaSO <sub>4</sub> : 218 mg L <sup>-1</sup> NaCl: 120 mg L <sup>-1</sup> NaNO <sub>3</sub> : 50 – 680 mg L <sup>-1</sup> MgCO <sub>3</sub> : 100 – 300 mg L <sup>-1</sup>	-0.8 to -1.4 V/SSE	Calcite Aragonite	<ul> <li>Flow rate</li> <li>Ca<sup>2+</sup> concentration</li> <li>Anions: SO<sub>4</sub><sup>2-</sup>, Cl<sup>-</sup>, NO<sub>3</sub><sup>-</sup></li> <li>Mg<sup>2+</sup></li> </ul>	[48]
Calcareous scaling	CaCO <sub>3</sub>	WE: Gold disk (0.2 cm²)  CE: Pt grid  RE: Saturated calomel electrode (SCE)	Carbonically pure water, $Ca^{2+}$ : $40-200$ mg $L^{-1}$ Dissolved $O_2$ : 2.4 to 4.8 mg $L^{-1}$ (mixture of $O_2$ and $N_2$ ), temperature: 25 and 50 °C, pH: 7	-1 V/SCE	Vaterite Spherulite Calcite	<ul> <li>Dissolved O<sub>2</sub> concentration</li> <li>Flow rate</li> <li>Degree of hardness</li> <li>Temperature</li> </ul>	[49]

Application	Precipitation	Electrochemical system	Electrolyte	Potential	Allotropy	Parameter	Ref
Calcareous scaling	CaCO <sub>3</sub>	WE: Gold disk (0.2 cm <sup>2</sup> )	Carbonically pure water  Ca <sup>2+</sup> : 40 and 200 mg L <sup>-1</sup> Dissolved O <sub>2</sub> : 5 or 100% (mixture of O <sub>2</sub> in N <sub>2</sub> ), flow rate: 60 and 1000 mL  min <sup>-1</sup>	-1.2 V/SCE	Calcite Vaterite	<ul> <li>Surface treatment</li> <li>Dissolved O<sub>2</sub>         concentration     </li> <li>Ca<sup>2+</sup> concentration</li> <li>Flow rate</li> </ul>	[50]
Calcareous scaling	CaCO <sub>3</sub>	WE: Soft steel, stainless steel or copper disk (1 cm²) CE: Tin oxide film coated conductive glass (transparent) RE: Ag-AgCl	Commercial mineral water, $Ca^{2+}: 108 \text{ mg L}^{-1}, Mg^{2+}: 14 \text{ mg L}^{-1}, \\ HCO_3^-: 350 \text{ mg L}^{-1}, Na^+: 3 \text{ mg L}^{-1}, \\ K^+: < 1 \text{ mg L}^{-1}, Cl^-: 9 \text{ mg L}^{-1}, SO_4^{2-}: \\ 61 \text{ mg L}^{-1}, pH: 7.4$	-1 V/Ag- AgCl	Calcite	<ul> <li>Nature of metallic substrate</li> <li>Surface energy of substrate</li> </ul>	[77]
Calcareous scaling	CaCO <sub>3</sub>	WE: Gold (0.2 cm²), bronze or stainless steel (1.5 cm²) CE: unspecified RE: Saturated copper sulfate electrode (SSE)	Carbonically pure water,  Ca <sup>2+</sup> : 40 and 200 mg L <sup>-1</sup> Simulated seawater,  CaCl <sub>2</sub> : 1.54 g L <sup>-1</sup> , NaHCO <sub>3</sub> : 168 mg  L <sup>-1</sup> , Na <sub>2</sub> CO <sub>3</sub> : 21 mg L <sup>-1</sup> , NaCl: 30 g  L <sup>-1</sup> , MgSO <sub>4</sub> : 3.48 g L <sup>-1</sup> , MgCl <sub>2</sub> : 8.31  g L <sup>-1</sup> , pH: 7 – 8	-0.9 to -1 V/SSE	Vaterite  Calcite  Aragonite	<ul> <li>Composition of scaling water</li> <li>Impedance study at different water composition</li> <li>Electrode substrate</li> </ul>	[78]
Calcareous scaling	CaCO <sub>3</sub>	WE: Carbon steel disk (0.2 cm²) CE: Pt plate RE: Saturated calomel electrode (SCE)	Synthetic solution corresponding to underground natural water (Tunisia) $Ca^{2+}$ : 100 mg $L^{-1}$ , $CO_3^{2-}$ : 4.2 – 5.1 mg $L^{-1}$ , $SO_4^{2-}$ : 500 – 1500 mg $L^{-1}$	Open circuit (corrosion potential)	Calcite Fe <sub>2</sub> O <sub>3</sub>	• SO <sub>4</sub> <sup>2-</sup> concentration • Rotation speed	[54]
Calcareous scaling	CaCO <sub>3</sub> Mg(OH) <sub>2</sub>	WE: Carbon steel disk (0.2 cm²) CE: Pt plate RE: Saturated calomel electrode (SCE)	Synthetic solution corresponding to underground natural water (Tunisia)  Ca <sup>2+</sup> : 100 mg L <sup>-1</sup> , Cl <sup>-</sup> : 996 mg L <sup>-1</sup> ,  HCO <sub>3</sub> -/CO <sub>3</sub> <sup>2-</sup> : unspecified, Mg <sup>2+</sup> : 30  – 142 mg L <sup>-1</sup> , pH: 7.0 – 7.5	-0.9 to -1.1 V/SCE	Calcite Mixture aragonite calcite Aragonite	• Mg <sup>2+</sup> concentration	[79]

Application	Precipitation	<b>Electrochemical system</b>	Electrolyte	Potential	Allotropy	Parameter	Ref
Calcareous scaling	CaCO <sub>3</sub>	WE: Carbon steel disk (0.2 cm²) CE: Pt plate RE: Saturated calomel electrode (SCE)	Synthetic solution corresponding to underground natural water (Tunisia)  Ca <sup>2+</sup> : 400 mg L <sup>-1</sup> , Cl <sup>-</sup> : 1000 mg L <sup>-1</sup> ,  CO <sub>3</sub> <sup>2-</sup> : 150 mg L <sup>-1</sup> , SO <sub>4</sub> <sup>2-</sup> : 480 – 3842  mg L <sup>-1</sup> , pH: 7.0 – 7.5	-0.95 V/SCE	Calcite Vaterite	• SO <sub>4</sub> <sup>2-</sup> concentration	[80]
Calcareous scaling	CaCO <sub>3</sub>	WE: Carbon steel disk (0.2 cm²) CE: unspecified RE: Saturated calomel electrode (SCE)	Natural underground water, Ca <sup>2+</sup> : 606 mg L <sup>-1</sup> , Mg <sup>2+</sup> : 279 mg L <sup>-1</sup> , HCO <sub>3</sub> <sup>-</sup> : 133 mg L <sup>-1</sup> , Na <sup>+</sup> : 1.3 g L <sup>-1</sup> , K <sup>+</sup> : 24 mg L <sup>-1</sup> , Cl <sup>-</sup> : 2.3 g L <sup>-1</sup> , SO <sub>4</sub> <sup>2-</sup> : 2.3 g L <sup>-1</sup> , Si: 5.8 mg L <sup>-1</sup> , pH: 7.2	Open circuit (corrosion potential)	-	<ul><li>Long immersion time</li><li>Rotation speed</li></ul>	[55]
Calcareous scaling	CaCO <sub>3</sub>	WE: Carbon steel disk (0.5 cm²) CE: Graphite bar RE: Saturated calomel electrode (SCE)	Simulated cooling water in refineries cooling towers  Ca <sup>2+</sup> :120 (low) or 360 (high) mg L <sup>-1</sup> ,  CO <sub>3</sub> <sup>2-</sup> : 180 (low) or 540 (high) mg L <sup>-1</sup> , Mg <sup>2+</sup> : 37 mg L <sup>-1</sup> , SiO <sub>2</sub> : 50 mg L <sup>-1</sup> ,  SO <sub>4</sub> <sup>2-</sup> : 200 mg L <sup>-1</sup> , Cl <sup>-</sup> : 200 mg L <sup>-1</sup> ,  PO <sub>4</sub> <sup>3-</sup> : 10 mg L <sup>-1</sup> , Fe <sup>2+</sup> : 0.1 mg L <sup>-1</sup> pH: 7.8, temperature: 40 °C	-1.1 to -1.75 V/SCE	Aragonite  Mixture aragonite calcite  Calcite	<ul> <li>Applied potential</li> <li>Concentrations of Ca<sup>2+</sup> and CO<sub>3</sub><sup>2-</sup></li> <li>Role of existing scale deposition on CaCO<sub>3</sub> crystals growth</li> </ul>	[81]
Calcareous scaling	CaCO <sub>3</sub>	WE: Carbon steel disk (0.5 cm²) CE: Graphite bar RE: Saturated calomel electrode (SCE)	Simulated cooling water in refineries cooling towers  Ca <sup>2+</sup> : 360 mg L <sup>-1</sup> , CO <sub>3</sub> <sup>2-</sup> : 540 mg L <sup>-1</sup> ,  Mg <sup>2+</sup> : 150 mg L <sup>-1</sup> , SiO <sub>2</sub> : 50 mg L <sup>-1</sup> ,  SO <sub>4</sub> <sup>2-</sup> : 200 mg L <sup>-1</sup> , Cl <sup>-</sup> : 200 mg L <sup>-1</sup> ,  PO <sub>4</sub> <sup>3-</sup> : 10 mg L <sup>-1</sup> , Fe <sup>2+</sup> : 0.1 mg L <sup>-1</sup> pH: 7.8, temperature: 40 °C	-1.1, -1.3 and -1.75 V/SCE	Aragonite  Mixture aragonite calcite  Calcite	• Role of existing scale deposition on CaCO <sub>3</sub> crystals growth	[82]

Application	Precipitation	<b>Electrochemical system</b>	Electrolyte	Potential	Allotropy	Parameter	Ref
Calcareous scaling	CaCO <sub>3</sub> Mg(OH) <sub>2</sub>	WE: Nickel foil or nickel film on Si substrate CE: Pt wire RE: Saturated calomel electrode	Synthetic industrial cooling water, CaCl <sub>2</sub> : 600 mg L <sup>-1</sup> , NaHCO <sub>3</sub> : 600 mg L <sup>-1</sup> , MgCl <sub>2</sub> : 300 mg L <sup>-1</sup> , organic inhibitor: 20 mg L <sup>-1</sup> either phosphonobutane tricarboxylic acid (PBTC) or ethane diphosphonic acid (EDPA), temperature: 25 – 65 °C	-0.5 to -0.9 V/SCE -0.8 to -1.4 V/SCE	Calcite Aragonite	<ul><li>Temperature</li><li>Water composition</li><li>Presence of organic inhibitors</li></ul>	[83]
Calcareous scaling	CaCO <sub>3</sub>	WE: Carbon steel disk (0.5 cm²) CE: Graphite bar RE: Saturated calomel electrode (SCE)	Simulated cooling water in refineries cooling towers  Ca <sup>2+</sup> :120 mg L <sup>-1</sup> , Mg <sup>2+</sup> : 150 mg L <sup>-1</sup> ,  SiO <sub>2</sub> : 50 mg L <sup>-1</sup> , SO <sub>4</sub> <sup>2-</sup> : 200 mg L <sup>-1</sup> ,  Cl <sup>-</sup> : 200 mg L <sup>-1</sup> , PO <sub>4</sub> <sup>3-</sup> : 10 mg L <sup>-1</sup> ,  Fe <sup>2+</sup> : 0.1 mg L <sup>-1</sup> ,  Corrosion inhibitor,  hydroxyphosphono acetic acid  (HPA): 3 mg L <sup>-1</sup> Scale inhibitor, hydroxyethane diphosphonic acid (HEDP): 3 mg L <sup>-1</sup> ,  pH: 7.8	Open circuit (corrosion potential)	No crystals but corrosion products	Presence of corrosion inhibitor, scale inhibitor or a mixture of both	[56]
Calcareous scaling	CaCO <sub>3</sub>	WE: Carbon steel disk (0.2 cm²) CE: Pt grid RE: Saturated calomel electrode (SCE)	Mineral potable water,  Ca <sup>2+</sup> : 112 mg L <sup>-1</sup> , Mg <sup>2+</sup> : 28 mg L <sup>-1</sup> ,  HCO <sub>3</sub> <sup>-</sup> : 430 mg L <sup>-1</sup> , Na <sup>+</sup> : 6.6 mg L <sup>-1</sup> ,  K <sup>+</sup> : 1.8 mg L <sup>-1</sup> , Cl <sup>-</sup> : 8 mg L <sup>-1</sup> , F <sup>-</sup> : 0.6  mg L <sup>-1</sup> , SO <sub>4</sub> <sup>2-</sup> : 61 mg L <sup>-1</sup> , NO <sub>3</sub> <sup>-</sup> : < 1  mg L <sup>-1</sup> , pH: 7.3, temperature: 40 °C  Green inhibitor: <i>Paronychia Argentea</i> (PA): 0 to 108 mg L <sup>-1</sup>	-1 V/SCE	Calcite  Deformed calcite	Presence of green inhibitor	[84]

Application	Precipitation	<b>Electrochemical system</b>	Electrolyte	Potential	Allotropy	Parameter	Ref
Calcareous scaling	CaCO <sub>3</sub>	WE: Stainless steel (1 cm²) CE: unspecified RE: unspecified	Simulated seawater, CaCl <sub>2</sub> : 28.45 g L <sup>-1</sup> , MgCl <sub>2</sub> : 5.75 g L <sup>-1</sup> , NaHCO <sub>3</sub> : 1.55 g L <sup>-1</sup> , NaCl: 63 – 64 g L <sup>-1</sup> , KCl: 2.25 g L <sup>-1</sup> , SrCl: 0.8 g L <sup>-1</sup> , inhibitors: 1 – 30 mg L <sup>-1</sup>	Unspecified	Vaterite  Aragonite  Calcite	• Impact of 3 green inhibitors and 1 nongreen inhibitor	[85]
Cathodic protection	CaCO <sub>3</sub>	WE: Gold disk (0.2 cm²) CE: Pt grid RE: Saturated calomel electrode (SCE)	Simulated seawater CaCl <sub>2</sub> : 1.2 g L <sup>-1</sup> , NaCl: 29 g L <sup>-1</sup> , NaHCO <sub>3</sub> : 168 mg L <sup>-1</sup> , Na <sub>2</sub> CO <sub>3</sub> : 21 mg L <sup>-1</sup> , pH: 8.2, stirring rate: 120 – 4900 rpm	-0.8 to -1.2 V/SCE	Calcite  Aragonite	Stirring rate	[57]
Cathodic protection	Mg(OH) <sub>2</sub>	WE: Gold disk (0.2 cm²)  CE: Pt grid  RE: Saturated calomel electrode (SCE)	Simulated seawater MgCl <sub>2</sub> : 5.2 g L <sup>-1</sup> , NaCl: 29 g L <sup>-1</sup> , NaHCO <sub>3</sub> : 168 mg L <sup>-1</sup> , Na <sub>2</sub> CO <sub>3</sub> : 21 mg L <sup>-1</sup> , pH: 8.2, stirring rate: 120 – 1200 rpm	-0.8 to -1.2 V/SCE	Brucite	• Stirring rate	[58]
Cathodic protection	CaCO <sub>3</sub> Mg(OH) <sub>2</sub>	WE: Carbon steel (2 cm²) and gold disk (0.2 cm²) CE: Pt grid RE: Saturated calomel electrode (SCE)	Simulated seawater CaCl <sub>2</sub> : 1.2 g L <sup>-1</sup> , MgCl <sub>2</sub> : 5.2 g L <sup>-1</sup> , NaCl: 25 g L <sup>-1</sup> , Na <sub>2</sub> SO <sub>4</sub> : 4.1 g L <sup>-1</sup> , NaHCO <sub>3</sub> : 234 mg L <sup>-1</sup> , KCl: 695 mg L <sup>-1</sup> , pH: 8.2, [Mg <sup>2+</sup> ]: 0 – 200% of [Mg <sup>2+</sup> ] <sub>0</sub>	-0.9 to -1.4 V/SCE	Aragonite Mixture aragonite brucite Brucite	<ul> <li>Metal substrate: steel and gold</li> <li>Mg<sup>2+</sup> concentration</li> <li>Applied potential</li> </ul>	[59]
Cathodic protection	CaCO <sub>3</sub> Mg(OH) <sub>2</sub>	WE: Carbon steel (2 cm²) CE: Pt grid RE: Saturated calomel electrode (SCE)	Simulated seawater  CaCl <sub>2</sub> : 1.2 g L <sup>-1</sup> , MgCl <sub>2</sub> : 5.2 g L <sup>-1</sup> ,  NaCl: 25 g L <sup>-1</sup> , Na <sub>2</sub> SO <sub>4</sub> : 4.1 g L <sup>-1</sup> ,  NaHCO <sub>3</sub> : 234 mg L <sup>-1</sup> , KCl: 695 mg  L <sup>-1</sup> , pH: 8.2, temperature: 10 – 30 °C,  SO <sub>4</sub> <sup>2-</sup> : 0 or 2.77 g L <sup>-1</sup> ,  [Ca <sup>2+</sup> ]: 0 – 100% of [Ca <sup>2+</sup> ] <sub>0</sub>	-1 and -1.2 V/SCE		<ul> <li>Temperature</li> <li>Ca<sup>2+</sup> concentration</li> <li>Presence of SO<sub>4</sub><sup>2-</sup></li> </ul>	[86]

Chapter II

Application	Precipitation	<b>Electrochemical system</b>	Electrolyte	Potential	Allotropy	Parameter	Ref
Cathodic protection	CaCO <sub>3</sub> Mg(OH) <sub>2</sub>	WE: Carbon steel (2 cm²) CE: Pt grid RE: Saturated calomel electrode (SCE)	Simulated seawater  CaCl <sub>2</sub> : 1.2 g L <sup>-1</sup> , MgCl <sub>2</sub> : 5.2 g L <sup>-1</sup> ,  NaCl: 25 g L <sup>-1</sup> , Na <sub>2</sub> SO <sub>4</sub> : 4.1 g L <sup>-1</sup> ,  NaHCO <sub>3</sub> : 234 mg L <sup>-1</sup> , 695 mg L <sup>-1</sup> ,  pH: 8.2, Ca <sup>2+</sup> : 0 or 421 mg L <sup>-1</sup> , Mg <sup>2+</sup> :  0 - 2.65 g L <sup>-1</sup> , SO <sub>4</sub> <sup>2-</sup> : 0 - 8.3 g L <sup>-1</sup>	-1 V/SCE	Aragonite  Brucite  Corrosion products	• SO <sub>4</sub> <sup>2-</sup> concentration • Mg <sup>2</sup> concentration	[87]
Cathodic protection	CaCO <sub>3</sub> Mg(OH) <sub>2</sub>	WE: Gold rod (0.2 cm²)  CE: Pt grid  RE: Saturated calomel electrode (SCE)	Simulated seawater  MgCl <sub>2</sub> : 5.2 g L <sup>-1</sup> , CaCl <sub>2</sub> : 1.2 g L <sup>-1</sup> ,  NaCl: 29 g L <sup>-1</sup> , NaHCO <sub>3</sub> : 168 mg L <sup>-1</sup> ,  Na <sub>2</sub> CO <sub>3</sub> : 21 mg L <sup>-1</sup> , pH: 8.2,  Na <sup>+</sup> -montmorillonite clay: 0 – 0.2 g  L <sup>-1</sup> , Kaolinite clay: 0 – 0.2 g L <sup>-1</sup>	-1.1 V/SCE	Aragonite  Calcite  Deformed calcite	Presence of clays in simulated seawater	[88]
Cathodic protection	CaCO <sub>3</sub>	WE: Transparent chromium-gold coated glass (0.4 cm²) CE: Pt grid RE: Saturated calomel electrode (SCE)	Carbonically pure water,  Ca <sup>2+</sup> : 160 mg L <sup>-1</sup> pH: 6.6	-1 V/SCE	Calcite Vaterite	-	[89]
Cathodic protection	CaCO <sub>3</sub> Mg(OH) <sub>2</sub>	WE: Steel (10 cm <sup>2</sup> ) CE: Pt foil RE: Saturated calomel electrode (SCE)	Simulated seawater, NaCl: 41 g L <sup>-1</sup> , NaHCO <sub>3</sub> : 210 mg L <sup>-1</sup> , Na <sub>2</sub> SO <sub>4</sub> : 4 g L <sup>-1</sup> , CaCl <sub>2</sub> : 1.1 g L <sup>-1</sup> , MgCl <sub>2</sub> : 5.2 g L <sup>-1</sup> , temp.: 0 – 25 °C	-0.906 V/SCE	Aragonite  Calcite	<ul><li> Temperature</li><li> Water composition</li></ul>	[90]
Cathodic protection	CaCO <sub>3</sub> Mg(OH) <sub>2</sub>	WE: Cylindrical different steel samples (20 cm²) CE: Pt coated mesh RE: Saturated calomel electrode (SCE)	Real seawater (sand filtered), Flow rate: 20 to 380 mL min <sup>-1</sup>	-0.9 V/SCE	Aragonite	<ul> <li>Different steel substrates and surface roughness</li> <li>Long polarization time</li> </ul>	[91]

Chapter II

Application	Precipitation	Electrochemical system	Electrolyte	Potential	Allotropy	Parameter	Ref
Cathodic protection	CaCO <sub>3</sub> Mg(OH) <sub>2</sub>	WE: Stainless steel (0.785 cm²) CE: Pt grid RE: Ag-AgCl	Simulated seawater, CaCl <sub>2</sub> : 1.2 g L <sup>-1</sup> , MgCl <sub>2</sub> : 5.2 g L <sup>-1</sup> , NaCl: 25 g L <sup>-1</sup> , Na <sub>2</sub> SO <sub>4</sub> : 4 g L <sup>-1</sup> , NaHCO <sub>3</sub> : 235 mg L <sup>-1</sup> , KCl: 703 mg L <sup>-1</sup> , pH: 8.2	-0.8 to -1.2 V/Ag-AgCl	Brucite  Aragonite	• Temperature • Applied potential	[60]
Cathodic protection	CaCO <sub>3</sub> Mg(OH) <sub>2</sub>	WE: Carbon steel disk (1 cm²) CE: Pt RE: Saturated calomel electrode (SCE)	Simulated seawater, CaCl <sub>2</sub> : 1.2 g L <sup>-1</sup> , MgCl <sub>2</sub> : 5.2 g L <sup>-1</sup> , NaCl: 24.5 g L <sup>-1</sup> , Na <sub>2</sub> SO <sub>4</sub> : 4.1 g L <sup>-1</sup> , NaHCO <sub>3</sub> : 201 mg L <sup>-1</sup> , KCl: 700 mg L <sup>-1</sup> , pH: 8.2, rotational speed: 120 – 600 rpm	-0.9 to -1.3 V/SCE	Aragonite Brucite	<ul><li>Rotational speed</li><li>Applied potential</li></ul>	[61]
Cathodic protection	CaCO <sub>3</sub> Mg(OH) <sub>2</sub>	WE: Stainless steel disk (1 cm²) CE: Pt RE: saturated calomel electrode (SCE)	Simulated seawater, CaCl <sub>2</sub> : 1.2 g L <sup>-1</sup> , MgCl <sub>2</sub> : 5.2 g L <sup>-1</sup> , NaCl: 24.5 g L <sup>-1</sup> , Na <sub>2</sub> SO <sub>4</sub> : 4.1 g L <sup>-1</sup> , NaHCO <sub>3</sub> : 200 mg L <sup>-1</sup> , KCl: 700 mg L <sup>-1</sup> , pH: 8.2 SO <sub>4</sub> <sup>2-</sup> : 0 – 200% [SO <sub>4</sub> <sup>2-</sup> ] <sub>0</sub> Mg <sup>2+</sup> : 0 – 200% [Mg <sup>2+</sup> ] <sub>0</sub> Ca <sup>2+</sup> : 0 – 200% [Ca <sup>2+</sup> ] <sub>0</sub>	-1.1 V/SCE	Calcite Aragonite	<ul> <li>Water composition</li> <li>SO<sub>4</sub><sup>2-</sup> concentration</li> <li>Mg<sup>2+</sup> concentration</li> <li>Ca<sup>2+</sup> concentration</li> </ul>	[62]

Abbreviations: WE: working electrode, CE: counter electrode, RE: reference electrode, Pt: platinum, Ti: titanium, Si: silicon.

#### II.2.2. Electrolytic properties and composition

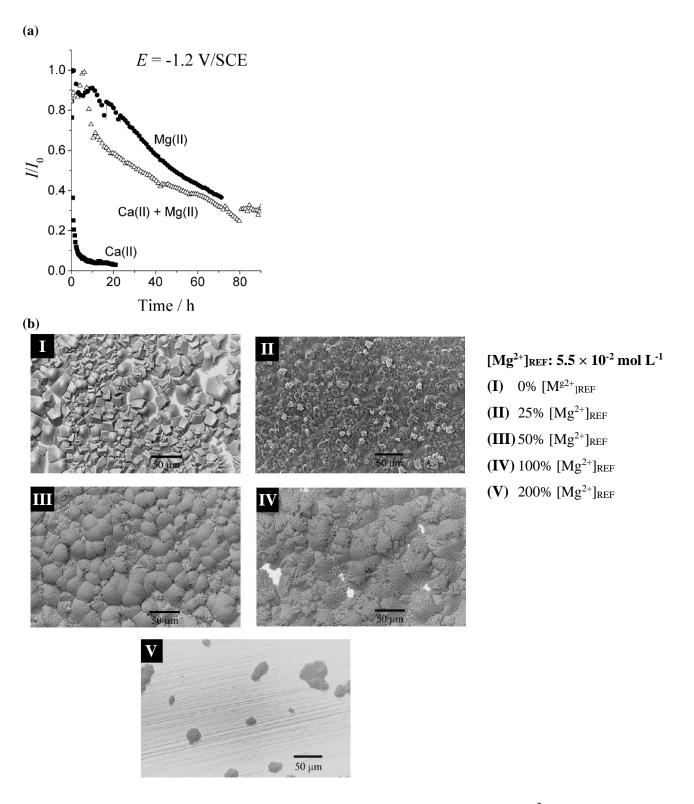
Simple Ca<sup>2+</sup> containing solution saturated by carbonate using CO<sub>2</sub> bubbling has been used in early stages to investigate calcareous kinetics and nuclei growth mechanism [46-48, 50, 78]. From there onwards, more complex matrices of solutions have been explored (Table II.4) such as simulated groundwater [54, 79, 80], simulated seawater [57-62, 85-87], simulated industrial cooling water [56, 81-83], natural groundwater [55] and real seawater [88, 90, 91] as well as mineral drinking water [77, 84]. This transition from simple matrix into real natural water matrix is necessary to investigate several possible parameters that could alter the kinetics of mineral scaling.

While treating real wastewater of multiple origins, the wastewater is often composed of multi-ions having various concentrations depending on their source. It means that several precipitations could occur once their thermodynamic constant of precipitation have been reached [92]. Consequently, cathodic precipitation is done deliberately to retain specific species from the bulk and the retained product could be recovered if it has an added value [93]. A clear example is the phosphorus nutrient recovery. Phosphorus is a nonrenewable resource and, in the meanwhile, the emission of phosphorus into the waste streams represents a rich unrecovered phosphorus source [94]. That is the motivation behind the phosphorus recovery via electrochemical precipitation method which has been attempted as electrochemical phosphorus recovery [95]. This technique is currently under optimization to replace chemical dosing method (e.g. with NaOH or Mg(OH)<sub>2</sub> to precipitate PO<sub>4</sub><sup>3-</sup> under the form of struvite (MgNH<sub>4</sub>H<sub>n</sub>PO<sub>4</sub><sup>3-n</sup>) in the waste streams rich with P and N nutrients) [93, 96]. A literature review about the recovery via electro-precipitation routes is provided in Table II.6.

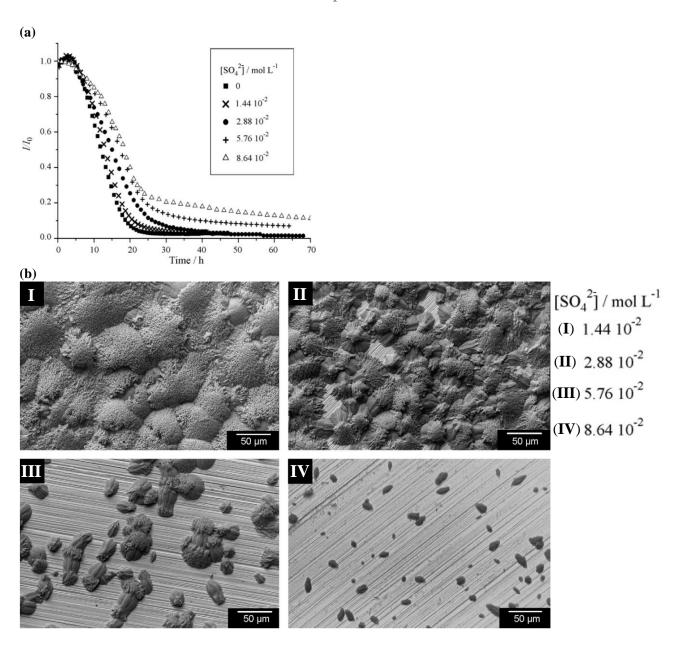
#### **II.2.2.1.** Presence of multi-ions in solution

Among the ionic species which are jointly present in the natural water body, Mg<sup>2+</sup> and SO<sub>4</sub><sup>2-</sup> are the two most reported species to inhibit the growth of scaling [59-62, 86]. Their inhibiting effect can be clearly seen from Figs. II.15-II.16 in which the chronoamperograms and scanning electron micrographs are respectively shown at varying concentrations of Mg<sup>2+</sup> and SO<sub>4</sub><sup>2-</sup>. Both chronoamperograms indicated an increase in residual currents as a result of lesser calcareous scaling on electrode surface. They were supported by scanning electron microscopy (SEM) images showing significantly bare electrode surface at the highest Mg<sup>2+</sup> and SO<sub>4</sub><sup>2-</sup> concentrations. According to literature, Mg<sup>2+</sup> hinders the growth of CaCO<sub>3</sub> by its incorporation into the CaCO<sub>3</sub> crystal lattice [79, 97, 98]. The growth of the most stable CaCO<sub>3</sub> crystal (calcite, cubic form as shown in Fig. II.15(b-I)) was then slowed down. It promoted aragonite crystallography (needle-like form as in Figs. II.15(b-II) to II.15(b-IV)) instead, as a result of incorporation of Mg<sup>2+</sup> into CaCO<sub>3</sub> deposit. As for

the SO<sub>4</sub><sup>2-</sup> ions, they could impede the CaCO<sub>3</sub> growth, either by lowering the cathode local pH [86, 87], adsorbing into the CaCO<sub>3</sub> crystal lattice [99], or by increasing the CaCO<sub>3</sub> solubility [62, 100].

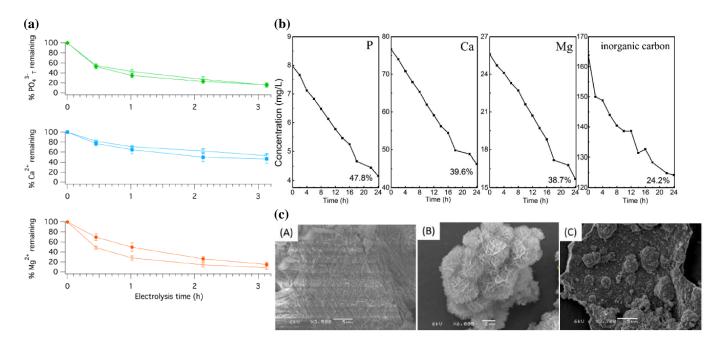


**Fig. II.15**. (a) Chronoamperometric curves in the absence and presence of Mg<sup>2+</sup> and (b) SEM of cathodic electro-precipitates at varying concentrations of Mg<sup>2+</sup>. Adapted from Barchiche et al. [59].



**Fig. II.16**. (a) Chronoamperometric curves in the presence of varying sulfate content and (b) SEM of cathodic electro-precipitates at the corresponding  $SO_4^{2-}$  concentration. Adapted from Barchiche et al. [87].

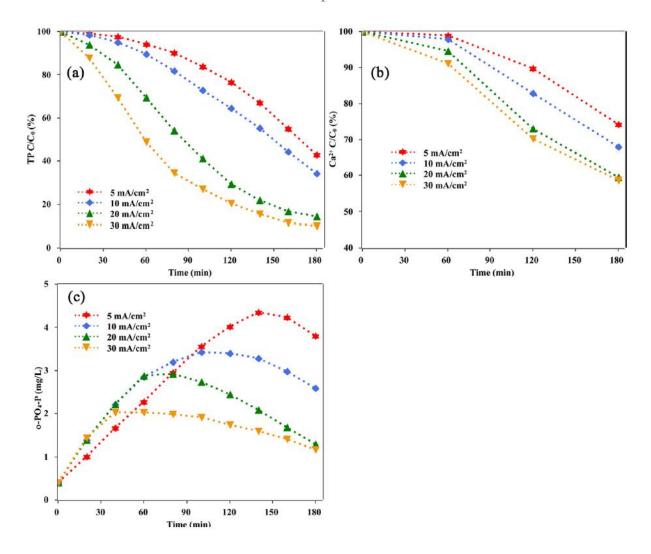
In the presence of other ionic species representative of typical raw [92, 96, 101, 102] and pretreated [103-106] wastewater, the occurrence of struvite (MgNH<sub>4</sub>H<sub>n</sub>PO<sub>4</sub><sup>3-n</sup>), calcium phosphates and other precipitates might occur. It has been documented that under the conditions where molar ratio of Mg<sup>2+</sup>, NH<sub>4</sub><sup>+</sup> and PO<sub>4</sub><sup>3-</sup> is adequate (1:1:1), struvite recovery via electro-precipitation was possible. Then, as depicted in Fig. II.17, when Ca<sup>2+</sup>, Mg<sup>2+</sup> and HPO<sub>4</sub><sup>2-</sup>/PO<sub>4</sub><sup>3-</sup> were simultaneously present in the effluent, the co-precipitation of Mg, Ca and phosphate might happen. Their concentrations decreased during the electrolysis treatment and it resulted in precipitates with regular morphologies and element distribution as shown in Fig. II.17(c). Where Ca was dominant, the precipitate took rhombohedral form (Fig. II.17(c-A)), while it presented lamella-like (Fig. II.17(c-B)) and spherical shapes (Fig. II.17(c-C)) when Mg and P element were predominantly present [92].



**Fig. II.17**. Evolution of PO<sub>4</sub><sup>3-</sup>, Ca<sup>2+</sup>, Mg<sup>2+</sup> and inorganic carbon during electro-precipitation process for their recovery when they were simultaneously present in the bulk. Treatment of (a) public wastewater reproduced from Cid et al. [101] and (b) influent of municipal wastewater treatment plant adapted from Lei et al. [92]. Their co-precipitation resulted in irregular morphologies as well as element distribution as shown from SEM images [92] in (c).

#### II.2.2.2. Presence of organics

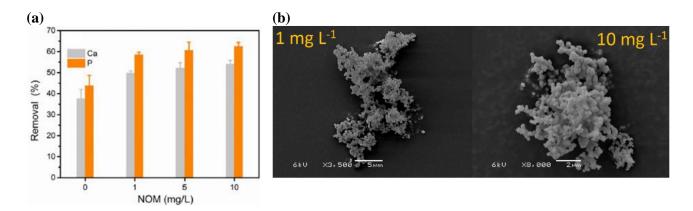
Electro-oxidation is applied to treat waste streams often containing organics and/or inorganic pollutants. The presence of organic compounds could somehow influence on the kinetics and the type of scaling formed on electrode surface. In the case where the organic pollutant itself is to be degraded (thus valueless) but it bears recoverable valuable resources, simultaneous organic abatement and resource recovery via electro-precipitation could be tempted. That was the essay conducted by Ning et al. [107] where organophosphorus pesticide was degraded via EO process using BDD on anode and simultaneous phosphorus recovery by cathodic precipitation in a single cell. Orthophosphate (PO<sub>4</sub><sup>3-</sup>) was released during the oxidation of the pesticide and it was later deposited on cathode via electro-precipitation with Ca<sup>2+</sup>. As a result, total phosphorus in the bulk decreased, PO<sub>4</sub><sup>3-</sup> increased and later decreased with Ca<sup>2+</sup> due to their co-precipitation on cathode (**Fig. II.18**). It could be deduced that depending on the composition of effluent to be treated, it could alter the degree of deposition occurring on anode or cathode.



**Fig. II.18**. Simultaneous electro-oxidation process of organophosphorus pesticide (acephate) on BDD anode and phosphorus recovery on Ti mesh cathode at different  $j_{app}$  [107]. The evolutions of (a) total phosphorus (TP), (b) Ca<sup>2+</sup> and (c) PO<sub>4</sub><sup>3-</sup> concentration during the treatment.

Anthropogenic dissolved organic matter (DOM) is omnipresent in wastewater and its content varies with anthropogenic activities and waste stream treatment level. When EAOP is to be used to treat wastewater as pretreatment or post-treatment, DOM will also be the subject of treatment in the electrochemical system. The interaction of DOM with precipitated compounds during an electrochemical recovery process has been systematically investigated in a millimetric reactor configuration [108, 109]. As opposed to the interaction of DOM with Ca<sub>3</sub>(PO<sub>4</sub>)<sub>2</sub> during chemical precipitation, more electro-precipitation of calcium phosphate was observed with increasing concentration of DOM during an electrochemical process as illustrated in Fig. II.19 [109]. It has been attributed to the rise of cathodic local pH that lifted the inhibition effect of DOM on calcium phosphate precipitation.

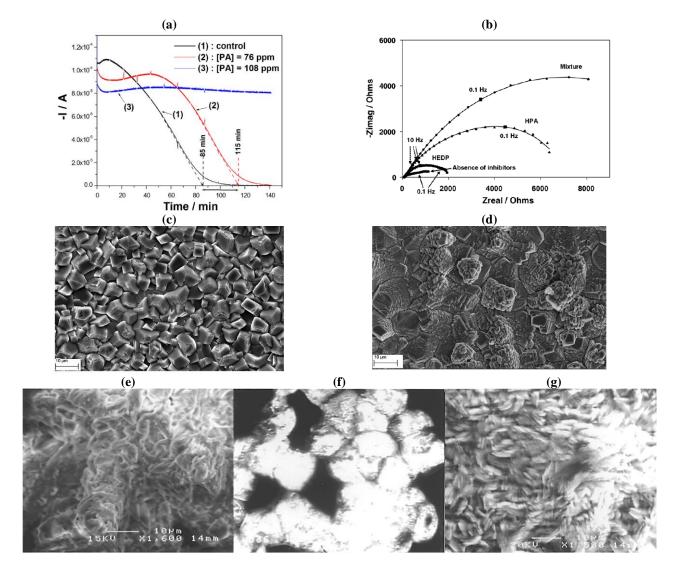
Consequently, the presence of DOM in the effluent could induce co-precipitation during an EAOP in general. Its behavior during the EO process under microfluidic condition using either synthetic or real wastewater will then be checked in the PhD thesis.



**Fig. II.19**. (a) Removal of Ca<sup>2+</sup> and phosphorus as a function of NOM concentration and (b) amorphous calcium phosphates precipitate. Adapted from Lei et al. [109].

#### **II.2.2.3.** Presence of scaling/corrosion inhibitors

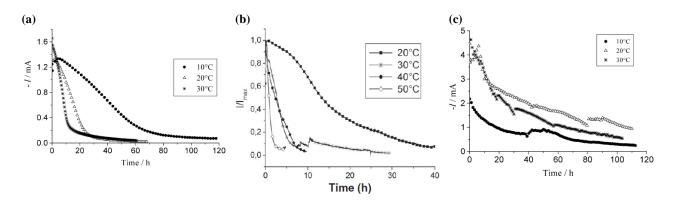
In the objective to reduce scaling tendency, the addition of inhibitors has been tested. Various studies have been devoted to the kinetic growth of CaCO<sub>3</sub> in the presence of inhibitors (**Table II.4**). The inhibitors could be either organic or inorganic compounds. As the name suggests, the growth of calcareous scaling was impeded in their presence. It can be affirmed by the plots of chronoamperometric and EIS curves in **Figs. II.20(a)** [84] and **II.20(b)** [56]. In the former, the curve did not decrease in the presence of optimal scale inhibitor (**Figs. II.20(a)**). In the latter, high charge transfer resistance ( $R_{CT}$ ) was measured (diameter of the demi-sphere) in the presence of inhibitors, which described significant corrosion activity of carbon steel (**Fig. II.20(b)**). Moreover, in the absence of scaling inhibitor, the CaCO<sub>3</sub> took typical cubic shape calcite crystallography (**Fig. II.20(c)** [84]). In the presence of inhibitors, deformed and irregular CaCO<sub>3</sub> morphologies have been documented (**Figs. II.20(d)-II.20(g)** [83, 84]). It has been proposed that the inhibitors inhibited the growth of calcareous scaling by adsorbing or co-precipitating with the calcareous layer [56, 83, 84].



**Fig. II.20**. (a) Chronoamperometric curves in the absence and presence of green inhibitor *Paronychia Argentea* [84], (b) EIS spectra during carbon steel corrosion in the presence of corrosion inhibitor (hydroxyphosphonoacetic acid), scale inhibitor (hydroxyethane diphosphonic acid) or both [56]. SEM of calcareous deposit in the absence of inhibitor (c) and in the presence of (d) PA inhibitor [84], (e) phosphonobutane tricarboxylic acid, (f) 1,1 ethane diphosphonic acid and (g) 1,2 ethane diphosphonic acid [83].

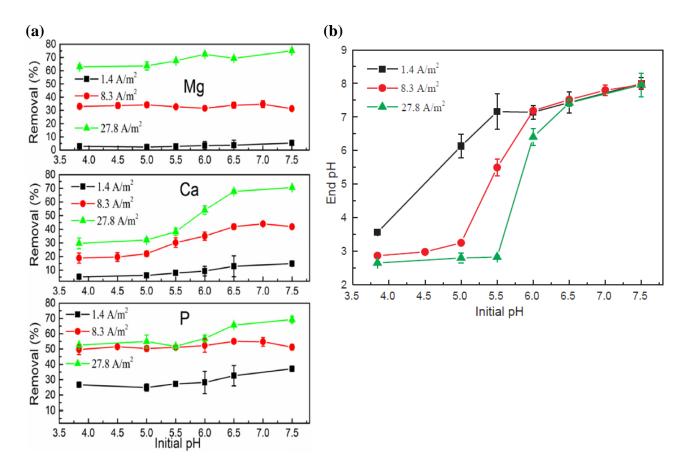
## II.2.2.4. Temperature and pH

The influence of water temperature has also been scrutinized, because it could have an impact on the solubility of calcareous scaling, dissolved O<sub>2</sub> solubility and diffusivity [60, 86, 90]. Figure II.21 illustrates the chronoamperograms during the formation of scaling at different solution temperatures and applied potentials. As the temperature increased, the solubility of CaCO<sub>3</sub> crystals decreased and the diffusivity of O<sub>2</sub> towards cathode increased [60, 86]. These two phenomenons explained the faster kinetics of deposition noticed in Figs. II.21(a) and II.21(b). Nonetheless, when a more cathodic potential was applied (-1.2 V/SCE), where H<sub>2</sub> evolution was significant, the influence of temperature was negligible (Fig. II.21(c)) [86].



**Fig. II.21**. Chronoamperograms during the formation of calcareous deposit at different solution temperatures. The experiments were performed at (a) -1.0 [86], (b) -1.1 [60] and (c) -1.2 V/SCE [86].

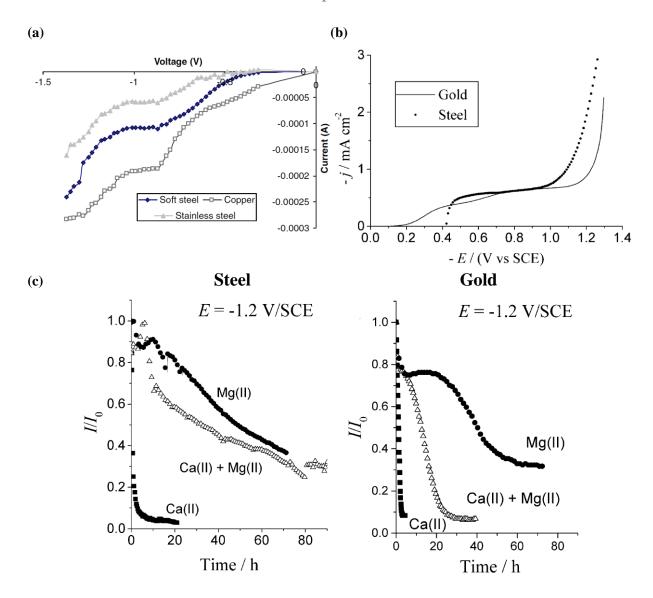
Bulk pH has primordial impact on the thermodynamic equilibrium of ionic species in the solution. EAOPs have different optimal operating pH notably those involving EF mechanism. According to Fig. II.22(a), the cathodic precipitation of Mg and phosphorus were less dependent on initial bulk pH, on the contrary to Ca precipitation [110]. It was ascribed to inorganic carbon speciation as function of pH, while alkaline pH promoted more Ca<sup>2+</sup> precipitation given the fact carbonate being the predominant form. It is also interesting to note that final effluent pH after electrochemical treatment is dependent on the initial pH of the effluent as schematized in Fig. II.22(b) [110]. Within the framework of EAOP treatment of reclaimed wastewater, the effluent pH is buffered by the presence of ionic species. Thus, the pH is often neutral.



**Fig. II.22**. (a) Precipitation of Mg, Ca and phosphorus as function of varying initial bulk pH and  $j_{app}$  as well as (b) dependency of final pH after electrolysis on initial bulk pH. Adapted from Lei et al. [110].

#### **II.2.3.** Electrode materials

Studies using different electrode materials as substrate for mineral scaling have been reported in literature and some of them have been listed in **Table II.4**. The electrode materials have diverse electrochemical responses under current/potential polarization. For example, as indicated in **Figs. II.23(a)-II.23(b)**, the voltammograms between electrode materials were distinct [59, 77]. From **Fig. II.23(a)**, faster O<sub>2</sub> reduction into OH<sup>-</sup> took place in respective order of copper, soft steel and stainless steel and the overpotential for HER were almost identical across all three [77]. In **Fig. II.23(b)**, O<sub>2</sub> reduction was firstly observed on gold but the potential window before H<sub>2</sub> evolution was larger on gold than on steel [59]. As a result, it was observed in **Fig. II.23(c)** that the scaling on gold was not yet perturbed by HER gas at similar applied potential compared with the steel [59]. Hence faster, more scaling and less residual current were noticed on gold.



**Fig. II.23**. (a, b) Voltammograms of different electrode materials polarized in cathodic region of synthetic solution containing precipitating elements [59, 77] and (c) chronoamperometric curves of precipitating Ca<sup>2+</sup> and/or Mg<sup>2+</sup> at different applied potentials [59]. Adapted from Euvrard et al. [77] and Barchiche et al. [59].

Elsewhere, carbonaceous graphite felt was compared to non-porous Pt-Ti electrode during the electroprecipitation process to remove and recover phosphorus (**Fig. II.24**) [111]. Better phosphorus deposition was noted on graphite felt in comparison to the Pt-Ti and it occurred with more acidic bulk pH. The authors have attributed the phenomenon towards higher accumulation of OH<sup>-</sup> in porous cathode resulting in higher precipitation.

Thus, porous cathode has indeed a crucial role on the electro-precipitation but yet little report has been made on this parameter. For that reason, it will be a parameter to investigate in the study of mineral scaling in submillimetric EAOP condition, as particularly done during the PhD thesis.

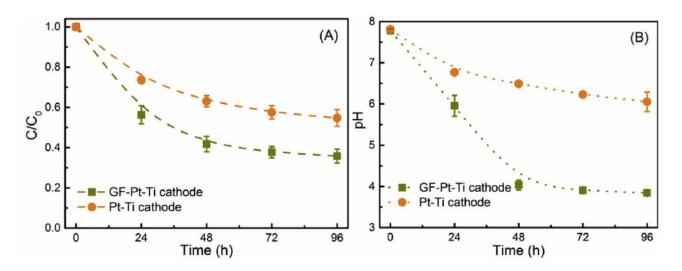


Fig. II.24. (a) Phosphorus concentration and (b) bulk pH evolution during electrochemical recovery using different cathode materials. Reproduced from Lei at. [111].

Table II.5. Struvite and phosphorus recovery as well as simultaneous pollutant removal using electrochemical recovery process.

Electro- precipitate	Electrochemical system	Effluent / electrolyte	$j_{ m app}$ or $E$	Parameters	Ref
Struvite (MgNH <sub>4</sub> PO <sub>4</sub> )	WE: Nickel rod (1 cm <sup>2</sup> ) CE: unidentified RE: Saturated Hg-HgSO <sub>4</sub> (SSE)	Representative of ionic properties of manure wastewater after sludge treatment NH <sub>4</sub> <sup>+</sup> : 69.5 mg L <sup>-1</sup> , H <sub>2</sub> PO <sub>4</sub> <sup>-</sup> /HPO <sub>4</sub> <sup>2-</sup> : 377.3 mg L <sup>-1</sup> , Mg <sup>2+</sup> : 93.6 mg L <sup>-1</sup>	-0.6 to -2 V/SSE	• Bulk pH	[103]
Struvite (MgNH <sub>4</sub> PO <sub>4</sub> )	Anode: Pt sheet (24 cm <sup>2</sup> ) Cathode: Nickel sheet (24 cm <sup>2</sup> ) or Pt sheet, Ru-Ir-alloy net (24 cm <sup>2</sup> ) and low porosity graphite rod ( $\varnothing$ 0.5 cm $\times$ 12 cm), $d_{\text{elec}}$ : 2 cm	Representative of ionic properties of waste stream from anaerobic digesters $PO_{4}^{3-}: 483 \text{ mg L}^{-1}, NH_{4}^{+}: 90 \text{ mg L}^{-1}, Mg^{2+}: 121 \text{ mg}$ $L^{-1}, pH: 7-7.5$	3 to 12 V	-	[104]
Struvite (MgNH <sub>4</sub> PO <sub>4</sub> )	WE: Stainless steel cylinder (5.39 cm²) CE: Pt grid RE: Ag-AgCl	Representative of ionic properties of animal manure $PO_4^{3-}$ : 760 mg $L^{-1}$ , $NH_4^+$ : 141 mg $L^{-1}$ , $Mg^{2+}$ : 260 mg $L^{-1}$ , $Ca^{2+}$ : 0 – 210 mg $L^{-1}$ , pH: 7.6, volume: 250 mL	-0.6 to -1.1 V/Ag-AgCl	<ul> <li>Applied potential</li> <li>Rotational speed</li> <li>Ca<sup>2+</sup> concentration</li> <li>Electrolyte temperature</li> </ul>	[105]
Phosphorus	Anode: BDD (37.5 cm <sup>2</sup> ) Cathode: Two Ti mesh (anode inserted in between) $d_{\text{elec}}$ : 1 cm	Synthetic solution, Acephate (organaphosphorus pesticide) giving out the total phosphorus content of: 10 mg L <sup>-1</sup> Na <sub>2</sub> SO <sub>4</sub> : 14 g L <sup>-1</sup> (supporting electrolyte), volume: 250 mL, pH: 4 – 9, Ca <sup>2+</sup> : 0 – 65 mg L <sup>-1</sup> , Mg <sup>2+</sup> : 0 or 25 mg L <sup>-1</sup> , HCO <sub>3</sub> -/CO <sub>3</sub> <sup>2-</sup> : 0 or 150 mg L <sup>-1</sup>	5 to 30 mA cm <sup>-2</sup>	<ul> <li>Applied current density</li> <li>Ca<sup>2+</sup> concentration</li> <li>Mg<sup>2+</sup> and coprecipitation of brucite</li> <li>HCO<sub>3</sub>- and coprecipitation of calcite</li> <li>Cl<sup>-</sup> (OH scavenger)</li> </ul>	[107]
Phosphorus	Divided cell, Anodic compartment: Pt (22 cm²), 200 mL Cathodic compartment: Steel (22 cm²), 200 mL	Nanofiltration concentrate from a dead-end filtration. This NF module treated the permeate of a pilot scale aerobic membrane bioreactor which was fed with domestic wastewater from local WWTP Ca <sup>2+</sup> : 83 mg L <sup>-1</sup> , Mg <sup>2+</sup> : 22 mg L <sup>-1</sup> , HCO <sub>3</sub> -/CO <sub>3</sub> <sup>2-</sup> : 58 mg L <sup>-1</sup> , PO <sub>4</sub> <sup>3-</sup> : 23 mg L <sup>-1</sup> , Na <sup>+</sup> : 130 mg L <sup>-1</sup> , K <sup>+</sup> : 22 mg L <sup>-1</sup> , pH: 7.8	0.46 mA cm <sup>-2</sup> (10 mA) 2.27 mA cm <sup>-2</sup> (50 mA)	<ul><li>Bulk pH</li><li>Applied current density</li></ul>	[106]

Electro- precipitate	Electrochemical system	Effluent / electrolyte	$j_{ m app}$ or $E$	Parameters	Ref
Phosphorus	Anode: MMO of Bi <sub>2</sub> O <sub>3</sub> -TiO <sub>2</sub> - Ir <sub>X</sub> Ta <sub>Y</sub> O <sub>2</sub> /Ti Cathode: Stainless steel Pilot scale: Volume: 22 L, recirculation flow rate: 10 L min <sup>-1</sup> , configuration: 7 anodes and 8 cathodes, separated by 3 mm distance. Surface area: 1.8 m <sup>2</sup> giving reactor surface to volume ratio of 80 m <sup>2</sup> m <sup>-3</sup> Bench scale configuration: Electrode spacing: 3 mm, reactor surface to volume ratio: 35 m <sup>2</sup> m <sup>-3</sup>	$\label{eq:constraints} Public toilet wastewater: $Ca^{2+}$: 40 mg L$^{-1}$, $Mg^{2+}$: 19.4 mg L$^{-1}$ $COD$: 320-380 mg-O$_2 L$^{-1}$, $HCO$_3$^-/CO$_3$^{2-}$: 1.04 g L$^{-1}$, $CI$^-$: 851 mg L$^{-1}$, $PO$_4$^3-$: 57 mg L$^{-1}$, $NH$_4$^+$: 235 mg L$^{-1}$, $SO$_4$^2-$: 288 mg L$^{-1}$, $Na$^+$: 621 mg L$^{-1}$, $K$^+$: 238 mg L$^{-1}$, pH: 8.3 $$Synthetic wastewater replicating the ionic and pH properties of public toilet waste water (for both pilot and bench scale): $$Ca^{2+}$: 6 - 58 mg L$^{-1}$, $Mg^{2+}$: 0 - 26 mg L$^{-1}$, $PO$_4$^3-$: 43 - 123 mg L$^{-1}$, $HCO$_3$^-/CO$_3$^2-$: 0.98 - 3.66 g L$^{-1}$$$	Pilot scale: Potentio static: 3.3 and 3.5 V  Bench scale: Potentio static: 3.5 V  Galvano static: 3 - 52 mA cm <sup>-2</sup>	wastewater  • Ca <sup>2+</sup> concentration  • Mg <sup>2+</sup> concentration  • PO <sub>4</sub> <sup>3-</sup> concentration  • HCO <sub>3</sub> -/CO <sub>3</sub> <sup>2-</sup> concentration	[101]
Phosphorus	Two cell compartments, 500 mL for electro-precipitation and another 500 mL for sampling mixing. Anode: Pt-coated Ti mesh Cathode: Ti plate (53 cm²)	Synthetic solution, P: 18.6 mg L <sup>-1</sup> , Ca <sup>2+</sup> : 40.1 mg L <sup>-1</sup> , Na <sub>2</sub> SO <sub>4</sub> : 7.1 g L <sup>-1</sup> , volume: 1000 mL (total of both cell compartments) recirculation flow rate: 100 mL min <sup>-1</sup> , pH: 4 - 10	0.4 mA cm <sup>-2</sup> (20 mA)	<ul><li>Bulk pH</li><li>Comparison with chemical precipitation method</li></ul>	[112]
Phosphorus	Two cell compartments, 500 mL for electro-precipitation and another 500 mL for sampling mixing. Anode: Pt-coated Ti mesh Cathode: Ti plate (53 cm <sup>2</sup> ) $d_{\rm elec}$ : 2.7 cm	Synthetic solution representative of domestic wastewater, $P: 6.2-46.5 \text{ mg L}^{-1}, \text{ Ca}^{2+}: 20 \text{ to } 100.2 \text{ mg L}^{-1}, \\ \text{Na}_2\text{SO}_4: 7.1 \text{ g L}^{-1}, \text{ pH}: 8.2 \\ \text{HCO}_3^{-7}\text{CO}_3^{2-}: 61-610 \text{ mg L}^{-1} \\ \text{Humic acid: } 0-10 \text{ mg L}^{-1} \\ \text{Volume: } 1000 \text{ mL}$	Open circuit and 0.19 to 1.89 mA cm <sup>-2</sup> (10 to 100 mA)	<ul> <li>Applied current density</li> <li>Ca/P ratio for phosphorus recovery</li> <li>HCO<sub>3</sub>-/CO<sub>3</sub><sup>2</sup>- concentration</li> <li>Presence of humic acid</li> </ul>	[108]

Electro- precipitate	Electrochemical system	Effluent / electrolyte	$j_{ m app}$ or $E$	Parameters	Ref
Phosphorus	Undivided cell, Anode: Ru-Ir coated Ti disk Cathode: Ti disk Surface area: 50 cm <sup>2</sup>	Synthetic solution,  Ca <sup>2+</sup> : 40 mg L <sup>-1</sup> , P: 7.4 mg L <sup>-1</sup> , Na <sub>2</sub> SO <sub>4</sub> : 7.1 g L <sup>-1</sup> ,  volume: 800 mL, pH: 4–10, flow rate: 70 mL min <sup>-1</sup> ,  Natural organic matter: 1 – 10 mg L <sup>-1</sup> NOM <sub>1</sub> : natural anionic polyelectrolyte, NOM <sub>2</sub> :  mixture of hydrophilic non-humic with  hydrophobic humic matter  NOM <sub>3</sub> : representative of fuvic acid	0.4 mA cm <sup>-2</sup> (20 mA)	<ul> <li>Type and concentration of natural organic matter (NOM)</li> <li>Bulk pH</li> </ul>	[109]
Phosphorus, CaCO <sub>3</sub> , Mg(OH) <sub>2</sub>	Undivided cell, Anode: Pt-coated Ti disk (50 cm <sup>2</sup> ), Cathode: Square Ti (4, 16 and 36 cm <sup>2</sup> ), $d_{\text{elec}}$ : 3 cm	Real wastewater influent of domestic WWTP (Leeuwarden, the Netherlands),  Ca <sup>2+</sup> : 75 mg L <sup>-1</sup> , Mg <sup>2+</sup> : 27 mg L <sup>-1</sup> , organic carbon:  40 mg L <sup>-1</sup> , inorganic carbon: 137 mg L <sup>-1</sup> , Cl <sup>-</sup> : 338 mg L <sup>-1</sup> , PO <sub>4</sub> <sup>3-</sup> : 7.7 mg L <sup>-1</sup> , NH <sub>4</sub> <sup>+</sup> : 66 mg L <sup>-1</sup> , SO <sub>4</sub> <sup>2-</sup> :  30 mg L <sup>-1</sup> , Na <sup>+</sup> : 299 mg L <sup>-1</sup> , K <sup>+</sup> : 24 mg L <sup>-1</sup> , NO <sub>3</sub> <sup>-1</sup> and NO <sub>2</sub> <sup>-</sup> : each < 0.1 mg L <sup>-1</sup> , pH: 8.3	5, 30 and 100 mA	<ul> <li>Applied current</li> <li>Cathodic surface area / applied current density</li> </ul>	[92]
Phosphorus, CaCO <sub>3</sub> , Mg(OH) <sub>2</sub>	Undivided cell, Anode: Pt-coated Ti disk (50 cm <sup>2</sup> ), Cathode: Square Ti (36 cm <sup>2</sup> ), $d_{\text{elec}}$ : 3 cm	Real wastewater influent of domestic WWTP (Leeuwarden, the Netherlands),  Ca <sup>2+</sup> : 81 mg L <sup>-1</sup> , Mg <sup>2+</sup> : 19 mg L <sup>-1</sup> , inorganic carbon:  138 mg L <sup>-1</sup> , Cl <sup>-</sup> : 223 mg L <sup>-1</sup> , PO <sub>4</sub> <sup>3-</sup> : 7.5 mg L <sup>-1</sup> ,  NH <sub>4</sub> +: 59 mg L <sup>-1</sup> , SO <sub>4</sub> <sup>2-</sup> : 26 mg L <sup>-1</sup> , Na <sup>+</sup> : 253 mg L <sup>-1</sup> ,  K <sup>+</sup> : 21 mg L <sup>-1</sup> , pH: 7.5, volume: 1000 mL, flow rate: 60 mL min <sup>-1</sup> , pH: 3.8 – 7.5 (HNO <sub>3</sub> )	5, 30 and 100 mA (0.14, 0.83 and 2.78 mA cm <sup>-2</sup> )	<ul><li>Bulk pH</li><li>Applied current density</li></ul>	[110]

Electro- precipitate	Electrochemical system	Effluent / electrolyte	$j_{ m app}$ or $E$	Parameters	Ref
Phosphorus, CaCO <sub>3</sub> , Mg(OH) <sub>2</sub>	Undivided cell batch and continuous cell, Anode: Pt-coated Ti disk (50 cm <sup>2</sup> ), Cathode: Square Ti or stainless steel (36 cm <sup>2</sup> ) $d_{\text{elec}}$ : 0.5, 3 and 6 cm	Real wastewater influent of domestic WWTP  (Leeuwarden, the Netherlands),  Ca <sup>2+</sup> : 75 mg L <sup>-1</sup> , Mg <sup>2+</sup> : 27 mg L <sup>-1</sup> , organic carbon:  40 mg L <sup>-1</sup> , inorganic carbon: 137 mg L <sup>-1</sup> , Cl <sup>-</sup> : 338 mg L <sup>-1</sup> , PO <sub>4</sub> <sup>3-</sup> : 7.7 mg L <sup>-1</sup> , NH <sub>4</sub> <sup>+</sup> : 66 mg L <sup>-1</sup> , SO <sub>4</sub> <sup>2-</sup> :  30 mg L <sup>-1</sup> , Na <sup>+</sup> : 299 mg L <sup>-1</sup> , K <sup>+</sup> : 24 mg L <sup>-1</sup> , NO <sub>3</sub> <sup>-1</sup> and NO <sub>2</sub> <sup>-</sup> : each < 0.1 mg L <sup>-1</sup> , pH: 8.3  Volume: 1000 mL and 1.2 mL min <sup>-1</sup> (both batch and continuous mode), electrolysis of 1 to 4 days	5, 30 and 100 mA (0.14, 0.83 and 2.78 mA cm <sup>-2</sup> )	<ul> <li>Applied current</li> <li>Electrolysis time, including long-term experiments</li> <li>Interelectrode distance</li> </ul>	[113]
Phosphorus	Microbial electrolysis cell, WE: Bio-anode, i.e., inoculated electroactive bacteria on graphite felt connected to Pt-coated Ti disk (50 cm²) CE: Ti plate (36 cm²), RE: Ag-AgCl delec: 1 cm NaAc substrate replacement every 24 h	Inoculation synthetic wastewater:  NaCH <sub>3</sub> COO (NaAc): 0.82 g L <sup>-1</sup> , KCl: 0.74 g L <sup>-1</sup> ,  NaCl: 0.58 g L <sup>-1</sup> , KH <sub>2</sub> PO <sub>4</sub> : 3.4 g L <sup>-1</sup> , K <sub>2</sub> HPO <sub>4</sub> : 4.35 g L <sup>-1</sup> , NH <sub>4</sub> Cl: 0.28 g L <sup>-1</sup> , MgSO <sub>4</sub> : 0.1 g L <sup>-1</sup> , CaCl <sub>2</sub> :  0.1 g/L  Biomass collected from active microbial recharge battery cell.  Real wastewater influent of domestic WWTP  (Leeuwarden, the Netherlands),  Ca <sup>2+</sup> : 81 mg L <sup>-1</sup> , Mg <sup>2+</sup> : 20 mg L <sup>-1</sup> , COD: 236 mg L <sup>-1</sup> , inorganic carbon: 121 mg L <sup>-1</sup> , organic carbon: 72 mg L <sup>-1</sup> , Cl <sup>-</sup> : 213 mg L <sup>-1</sup> , PO <sub>4</sub> <sup>3-</sup> : 7 mg L <sup>-1</sup> , NH <sub>4</sub> <sup>+</sup> : 52 mg L <sup>-1</sup> , SO <sub>4</sub> <sup>2-</sup> : 27 mg L <sup>-1</sup> , Na <sup>+</sup> : 249 mg L <sup>-1</sup> , K <sup>+</sup> : 21 mg L <sup>-1</sup> , pH: 7.5, volume: 900 mL	-0.35 V/Ag-AgCl (potential slightly higher than the oxidation potential of NaAc substrate)	<ul> <li>NaAC substrate concentration</li> <li>Operation duration</li> <li>Initial phosphate concentration</li> </ul>	[114]

Electro- precipitate	Electrochemical system	Effluent / electrolyte	$j_{ m app}$ or $E$	Parameters	Ref
Dhoonhome	Column electrochemical reactor containing CaCO <sub>3</sub> particles, Anode: Pt-coated Ti rod (14 cm <sup>2</sup> ), Cathode: Ti rod (14 cm <sup>2</sup> ),	Synthetic solution, P: 18.6 mg L <sup>-1</sup> and 1 mg L <sup>-1</sup> (for low P system) Na <sub>2</sub> SO <sub>4</sub> : 1.42 g L <sup>-1</sup>	5, 10 and 20 mA (0.36, 0.71	<ul> <li>Applied current density</li> <li>CaCO<sub>3</sub> particle sizes</li> <li>Batch vs. continuous mode</li> </ul>	[115]
Phosphorus	$d_{\rm elec}$ : 1 cm Continuous flow mode CaCO <sub>3</sub> particle size: < 0.5, 0.5-1, 1-2 and 2-3 mm	Real wastewater influent of domestic WWTP (Leeuwarden, the Netherlands) Flow rate: $0.4 - 2.4 \text{ L day}^{-1}$	and 1.43 mA cm <sup>-2</sup> )	<ul> <li>Initial P content</li> <li>Synthetic vs real wastewater</li> <li>Long term test</li> </ul>	[115]
Phosphorus	Undivided cell, Anode: Pt-coated Ti disk (50 cm <sup>2</sup> ) Cathode: Graphite felt (50 cm <sup>2</sup> ) attached to Pt-Ti disk $d_{\text{elec}}$ : 3 cm	Synthetic solution, P: 18.6 mg L <sup>-1</sup> , Ca <sup>2+</sup> : 20 – 80.2 mg L <sup>-1</sup> , Na <sub>2</sub> SO <sub>4</sub> : 1.42 – 7.1 g L <sup>-1</sup> Real wastewater influent of domestic WWTP (Leeuwarden, the Netherlands) Volume: 900 mL, flow rate: 120 mL min <sup>-1</sup>	0.2 mA (low) 1 mA (highest)	<ul> <li>Applied current</li> <li>Initial Ca<sup>2+</sup> concentration</li> <li>Synthetic vs. real wastewater</li> </ul>	[111]
Phosphorus	Undivided cell, Anode: Pt-coated nickel, Ru-Ir MMO or Pt-Ir MMO (50 cm²) Cathode: Ti square plate (36 cm²) $d_{\text{elec}} : 3 \text{ cm}$	Synthetic solution, Ca <sup>2+</sup> : 40 mg L <sup>-1</sup> , Na <sub>2</sub> SO <sub>4</sub> : 7.1 g L <sup>-1</sup> (50 mM) or NaCl: 50 mM or NaClO <sub>4</sub> : 50 mM, Nitrilotris (methylene phosphonic acid) (NTMP) (non- orthophosphates source): 30 mg L <sup>-1</sup> (0.3 mM P), Volume: 1000 mL Real wastewater influent of domestic WWTP (Leeuwarden, the Netherlands) spiked with 30 mg L <sup>-1</sup> NTMP	2.8 mA cm <sup>-2</sup> (100 mA)	<ul><li>Anode material</li><li>Electrolyte composition</li><li>Synthetic vs. real wastewater</li></ul>	[116]

Chapter II

Electro- precipitate	Electrochemical system	Effluent / electrolyte	$j_{ m app}$ or $E$	Parameters	Ref
Phosphorus	Undivided cell, Anode: Pt-coated Ti (50 cm <sup>2</sup> ), Cathode: Pt-coated Ti (50 cm <sup>2</sup> ), $d_{\text{elec}}$ : 3 cm	Raw cheese wastewater from industrial WWTP  (Groningen, the Netherlands)  Ca <sup>2+</sup> : 2.2 g L <sup>-1</sup> , Mg <sup>2+</sup> : 138 mg L <sup>-1</sup> , COD: 19.8 g L <sup>-1</sup> ,  soluble P: 827 mg L <sup>-1</sup> , total carbon: 7.72 g L <sup>-1</sup> ,  organic carbon: 7.71 g L <sup>-1</sup> , Cl <sup>-</sup> : 107 mg L <sup>-1</sup> , PO <sub>4</sub> <sup>3-</sup> :  2.4 g L <sup>-1</sup> , NH <sub>4</sub> <sup>+</sup> : 77 mg L <sup>-1</sup> , total N: 496 mg L <sup>-1</sup> ,  SO <sub>4</sub> <sup>2-</sup> : 257 mg L <sup>-1</sup> , Na <sup>+</sup> : 87 g L <sup>-1</sup> , K <sup>+</sup> : 21 mg L <sup>-1</sup> , pH:  4.4, conductivity: 209 mS cm <sup>-1</sup> , volume: 900 mL,  flow rate: 150 mL min <sup>-1</sup>	1 to 6 mA cm- 2 (50 to 300 mA)	<ul> <li>Chemical (NaOH dosage) vs.</li> <li>electrochemical precipitation tests</li> <li>Bulk pH</li> <li>Applied current density</li> </ul>	[96]
Phosphorus	Scale-up study Prototype 1  Volume: 4 L, Anode: Ru-Ir MMO placed in the center of reactor Cathode: Tubular stainless steel positioned adjacent to the inner wall of reactor Prototype 2  Volume: 2.8 L, Anode: Pt-coated Ti, Cathode: Tubular stainless steel with 35% less surface area (economical reason). Closed system, H <sub>2</sub> (cathode) and Cl <sub>2</sub> (anode) were sent to gas collector (1 M NaOH) Flow rate: 0.93 L day <sup>-1</sup> (0.65 mL min <sup>-1</sup> ) in long run test	Raw cheese wastewater from industrial WWTP (Groningen, the Netherlands)  Prototype 1  Parametric investigation: Retention time: 1, 1.5 and 3 days P concentration: 1×, 2× and 10× dilution factor  Prototype 2  Long term experiment  Applied current: 600 mA, retention time: 3-day, flow rate: 0.93 L day-1 (0.65 mL min-1)	200 – 800 mA	<ul> <li>Applied current</li> <li>Residence time, including long-term experiments</li> <li>Phosphate concentration</li> </ul>	[102]

<u>Abbreviations</u>: BDD: boron-doped diamond, MMO: mixed metal oxide, Pt: platinum, Ti: titanium, Ir: iridium, Ru: ruthenium, COD: chemical oxygen demand, P: phosphorus,  $d_{\text{elec}}$ : interelectrode distance and  $j_{\text{app}}$ : applied current density.

### II.2.4. Electrochemical reactor design, scale up and maintenance

The electro-precipitation for nutrient recovery has been receiving increasing attention in recent years, owing to its promising possibility to recover valuable products. Therefore, several reactors have been designed at large scale to increase the treatment capacity as depicted in Figs. II.25(a)-II.25(c). Takabe et al. used a batch electrochemical reactor to precipitate phosphorus using 4-pairs of electrodes working in series to treat up to 5 L of effluent (Fig. II.25(a)) [117]. Lei et al. started the parametric studies in a reactor up to 1 L capacity using Pt-coated Ti electrodes having 50 cm<sup>2</sup> surface area [92, 109, 113] (Fig. II.25(b)). They recently came up with a pilot scale reactor unit with tubular stainless steel cathode with a capacity to treat up to 4 L in continuous mode (Fig. II.25(c)) [102]. Here, after long operation hours, the cathode was totally/almost passivated by the precipitates. As a countermeasure, Takabe et al. operated their reactors with polarity reversal to improve the process durability.

In regard to mineral scaling, most of the studies listed in Table II.4 involve the use of a classical rotating disk electrode with varying exposed surface area up to 2 cm<sup>2</sup>. The use of such electrodes is only adequate to conduct mechanistic studies on the kinetics of precipitation. However, the fundamental of scaling deposition remains valid, while operating the electrolysis to treat wastewater at an industrial scale, for example in water softening technology. Due to the precipitation on the cathode, Mg- and Ca-based precipitates were formed and the water was softened. Water softening technique has been widely adopted to treat a large spectrum of effluent according to the list provided in Table II.6. The electro-precipitation concept for water softening has received huge success and it has been employed from laboratory up to big scale industrial projects. For example, it has been applied in a case study where a commercial filter-press reactor (Fig. II.25(d)) [118] and a multistage electrochemical precipitation reactor (Fig. II.25(e)) [119] have been employed to induce mineral scaling for water softening purpose. Elsewhere, at relatively larger scale, water hardness, COD and total nitrogen (TN) of the secondary effluent of a nonradioactive WWTP of nuclear power plant (South Korea) were treated using integrated electrodialysis reversal coupled to EO process [120]. Another one involved water hardness and trihalomethanes treatment of brackish Llobregat River (Barcelona, Spain) by electrodialysis with polarity reversal technology [121].

These successful industrial scale water treatment and water softening have only been made possible owing to the implementation of periodically self-cleaning electrodes and/or membranes so that the total passivation of electrodes did not occur. For that reason, the most realistic, cost effective and on-the-field proven technique is the periodic reversal. However, this technique might not suit most of EAOPs. Efficient descaling using this technique could only be obtained by applying high applied

current (**Table II.6**), which means an increase of the overall operation cost. Furthermore, only stable electrodes should be used (refer **Table II.6**) such as DSA (IrO<sub>2</sub>-RuO<sub>2</sub>/Ti), MMO (IrO<sub>2</sub>-RuO<sub>2</sub>-SnO<sub>2</sub>/Ti), Pt-coated Ti or BDD that could resist high current polarization at both cathode and anode configurations. Certain electrode materials are corroded upon anodic polarization at high current, releasing ions into the solution (e.g., most steels and less noble metals). Carbonaceous materials are even burnt when they are used as anode under medium and high current densities. Therefore, one of the objectives of this thesis work is to check whether an alternative for polarity reversal technique would be possible.

Upon deep literature survey on the topic of cathodic electro-precipitation, there has not yet been a study that reports on the precipitation occurring inside submillimetric electrochemical reactors. It is indeed a step backward with respect to the scale up attempt aforementioned. Given the interesting advantages that could offer microfluidic reactors and the fact that they are susceptible for scale up (Section II.1.2), the scaling phenomenon inside submillimetric configuration represents a research gap. Only a couple of studies have attempted to investigate the role of  $d_{\rm elec}$  towards the kinetics of electro-precipitation. Lei et al. observed a decrease in precipitation of phosphorus going from 30 mm to 5 mm and they attributed it to the compensation of anodic H<sup>+</sup> on cathodic OH<sup>-</sup> thus lowering the precipitating capacity [113]. Takabe et al. went slightly below to 1 mm from 10 mm and 5 mm as they also noticed less electro-precipitation on cathode at the lowest investigated  $d_{\rm elec}$  [117]. However, they observed that the phosphorus precipitation followed a different trend with 1 mm  $d_{\rm elec}$ , whereby more precipitation on the bulk was noticed as a result of higher bulk pH. Therefore, an interesting observation has been made at lower  $d_{\rm elec}$ .

Since no approach has yet been attempted on the occurrence of electro-precipitation in microfluidic cells, this PhD thesis intends to serve as a reference for the sequencing works in the future. The involvement of anodic activity would be supposedly more marked and it might play significant role towards the mechanism of electro-precipitation on cathode as well as pollutant degradation via BDD(\*OH) or mediated oxidants in the bulk in-between electrodes. Moreover, the kinetic model to estimate and predict mineral scaling is lacking in the current literature. Hence, it will be another quest invoked in this thesis.

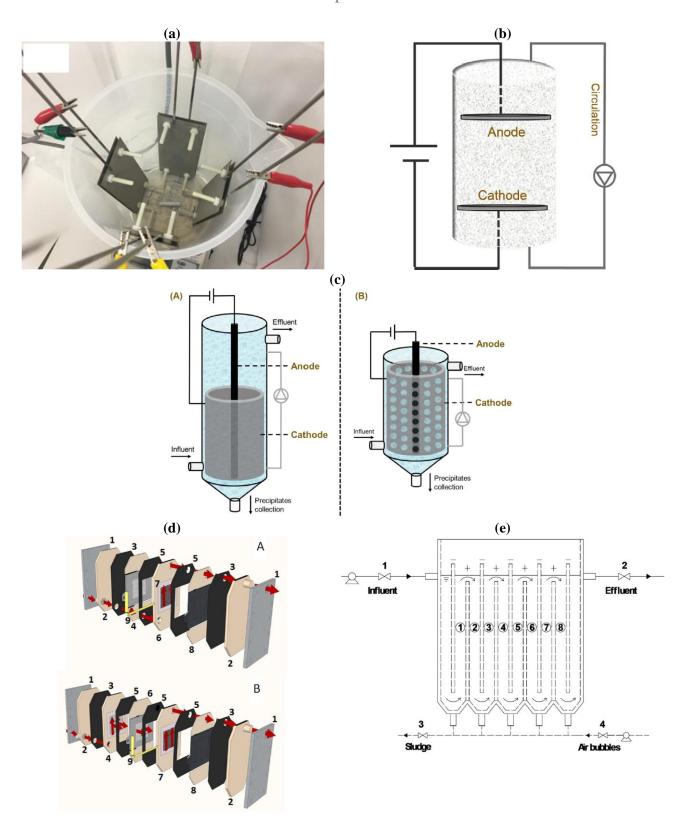


Fig. II.25. (a) 4-pair of electrodes working in series by Takabe et al. [117], (b) horizontal parallel-plate reactor [116] and (c) scaled up electrochemical reactor with large tubular cathode for phosphorus electro-precipitation recovery [102] adapted from Lei et al., (d) commercial filter-press type reactor by Sanjuán et al. [118] and (e) multistage electro-precipitation reactor for water softening technology by Yu et al. [119].

Table II.6. Electro-precipitation for water softening technology and successive implementation of polarity reversal to minimize total passivation.

Process	Electrochemical system	Effluent / electrolyte	$j_{ m app}$ or $E$	Parameters	Ref
Water softening Polarity reversal	Electro-deionization with periodic polarity change (EDIR)  6 pairs of EDIR stacks consisting of cationic and anionic exchange membranes in 3 stages  Anode = Cathode = Ti-coated Pt  Surface area: 100 cm <sup>2</sup>	Synthetic solution, Total hardness: 250 – 1250 mg-CaCO <sub>3</sub> L <sup>-1</sup> Flow concentrate: 25 – 500 mL min <sup>-1</sup> Flow diluate: 20 mL min <sup>-1</sup> (constant) Polarity reversal: 10 – 60 min	20 V	<ul> <li>Flow rate</li> <li>Polarity reversal period</li> <li>Water hardness</li> <li>Comparison on long term operation using homogenous and heterogeneous membrane</li> </ul>	[122]
Removal of COD and TN Polarity reversal	Integrated electrodialysis reversal (EDR) with electro-oxidation (EO) processes  Electrodialysis reversal  2-stack in series each containing 250 cell pairs.  Each pair consisted of diluting compartment, cation membrane, concentrating compartment and anion membrane  Membrane area: 1440 cm²  Electro-oxidation  Bipolar electrolyzer, parallel flow, 23 electrodes of RuO <sub>2</sub> -SnO <sub>2</sub> -IrO <sub>2</sub> /Ti MMO, delec: 1.4 cm, effective surface area: 1800 cm²	Effluent of secondary nonradioactive WWTP of nuclear power plant (South Korea)  Turbidity: 0.57 NTU, pH: 7.78, conductivity: 2.5 mS cm <sup>-1</sup> , suspended solid: 35 mg L <sup>-1</sup> , COD: 33 mg L <sup>-1</sup> , TN: 36 mg L <sup>-1</sup> , Ca <sup>2+</sup> : 41 mg L <sup>-1</sup> , Mg <sup>2+</sup> : 66 mg L <sup>-1</sup> , Fe <sup>2+</sup> : 1.6 mg L <sup>-1</sup> , Al <sup>3+</sup> : 2.2 mg L <sup>-1</sup> , Mn <sup>2+</sup> : 0.06 mg L <sup>-1</sup> , Na <sup>+</sup> : 566 mg L <sup>-1</sup> , K <sup>+</sup> : 18 mg L <sup>-1</sup> , NH <sub>4</sub> <sup>+</sup> : 8.2 mg L <sup>-1</sup> , Cl <sup>-</sup> : 557 mg L <sup>-1</sup> , SO <sub>4</sub> <sup>2-</sup> : 695 mg L <sup>-1</sup> , NO <sub>3</sub> <sup>-</sup> : 0.05 mg L <sup>-1</sup> , SiO <sub>2</sub> : 2.7 mg L <sup>-1</sup> , total dissolved solid: 1.25 g L <sup>-1</sup>	EDR Product 13 L min <sup>-1</sup> Concentrate 2.8 L min <sup>-1</sup> Stage I 60 V – 77 A Stage II 60 V – 3.7 A  EO 6 L min <sup>-1</sup> 126 A (70 mA cm <sup>-2</sup> )	• Industrial application	[120]

Process	Electrochemical system	Effluent / electrolyte	$j_{ m app}$ or $E$	Parameters	Ref
Tri- halomethane (THM) removal and desalination  Polarity reversal	Electrodialysis with electrode polarity reversal (EDR) to automatically clean membrane surface as the tertiary treatment.  Cartridge filter of 10 µm before going into the EDR unit.  Electrode material: unspecified  Pilot plant  2-stage stacks with 200 cell pairs each stack, maximum DWTP flow of 3 m³ s⁻¹  Up-scale unit  Industrial pilot scale with 2 stacks of 600 cell each, maximum DWTP flow of 4 m³ s⁻¹	Brackish river water (Llobregat, Barcelona) after conventional treatment (pre-oxidation using potassium permanganate, coagulation, flocculation, oxidation with ClO <sub>2</sub> , sand filtration, granulated active carbon filtration and final chlorination using chlorine gas) at drinking water treatment plant (DWTP)	unspecified	• Industrial application	[121]
Scaling detachment using pulsed applied current	Flow plant with undivided cell, Anode: DSA sheet, cathode: mirror stainless steel Surface area: 147 cm <sup>2</sup> $d_{\text{elec}}$ : 5 mm	Simulated properties of industrial recirculated cooling water, CaCl <sub>2</sub> : 333 mg L <sup>-1</sup> , NaHCO <sub>3</sub> : 504 mg L <sup>-1</sup> , MgCl <sub>2</sub> : 101.5 mg L <sup>-1</sup> , anti-scalant: 2 mg L <sup>-1</sup> , pH: 8.5, conductivity: 1400 µs cm <sup>-1</sup> , V: 100 mL, Flow rate: 0.08 – 0.33 L min <sup>-1</sup>	10 to 60 mA cm <sup>-2</sup>	<ul><li>Water softening performance</li><li>Applied current density</li><li>Flow rate</li></ul>	[123]
Water softening	Multistage electrochemical precipitation reactor, 8-stage cell, Anode: $4 \times$ DSA, Cathode: $5 \times$ stainless steel, $d_{\rm elec}$ : 1 cm, total effective surface area: $800~{\rm cm}^2$ Single stage surface area: $100~{\rm cm}^2$	Simulated properties of industrial recirculated cooling water  Hardness: 350 mg L <sup>-1</sup> , alkalinity: 300 mg L <sup>-1</sup> , pH: 8.2, anti-scalant: 1 – 6 mg L <sup>-1</sup> Periodic scale detachment by sending air bubbles from the bottom of the cell	2 to 12 mA cm <sup>-2</sup> Single cell: 0.2 to 1.2 A Multistage: 1.6 to 9.6 A	<ul><li>Applied current density</li><li>Flow rate</li><li>Anti-scalant concentration</li></ul>	[119]
Water softening Polarity reversal	Flow plant with undivided cell, Anode = Cathode = $IrO_2$ -RuO <sub>2</sub> /Ti DSA (147 cm <sup>2</sup> ) $d_{elec}$ : 5 mm	Simulated properties of recirculating cooling water,  CaCl <sub>2</sub> : 333 mg L <sup>-1</sup> , NaHCO <sub>3</sub> : 504 mg L <sup>-1</sup> ,  MgCl <sub>2</sub> : 102 mg L <sup>-1</sup> , pH: 8.5, conductivity:  1600 µS cm <sup>-1</sup> , volume: 100 mL  Flow rate: 0.17 – 0.5 L min <sup>-1</sup>	10 to 25 mA cm <sup>-2</sup>	<ul><li>Polarity reversal time</li><li>Flow rate</li><li>Long cycle for stability test</li></ul>	[124]

Process	Electrochemical system	Effluent / electrolyte	$j_{ m app}$ or $E$	Parameters	Ref
Water softening Polarity reversal	Flow plant with undivided cell, Anode = Cathode = DSA (200 cm <sup>2</sup> ), 4 membranes (8 mm thickness each) and 5 plastic grids were placed in between electrodes	Simulated properties of recirculating cooling water,  CaCl <sub>2</sub> : 555 mg L <sup>-1</sup> , NaHCO <sub>3</sub> : 840 mg L <sup>-1</sup> Flow rate: 0.33 – 3 L min <sup>-1</sup>	10 mA cm <sup>-2</sup>	<ul> <li>Heterogeneous vs.         <ul> <li>homogenous exchange membrane</li> </ul> </li> <li>Ca<sup>2+</sup> concentration</li> <li>Mg<sup>2+</sup> concentration</li> <li>Alkalinity</li> <li>Ionic conductivity</li> <li>Membrane spacing</li> <li>Scale detachment and operation stability tests</li> </ul>	[125]
Phosphorus recovery Polarity reversal	Electrochemical batch reactor, Anode = Cathode = Pt-coated Ti 2 pairs of electrodes were used connected in parallel $d_{\text{elec}}$ : 1 cm Polarity inversion period: 1 – 10 min	Real anaerobic digestion effluent pretreated with cationic organic polymer coagulant and centrifugally dehydrated  Total phosphorus: 74 mg L <sup>-1</sup> , dissolved phosphorus: 57 mg L <sup>-1</sup> , PO <sub>4</sub> <sup>3-</sup> : 51 mg L <sup>-1</sup> , Ca <sup>2+</sup> : 44 mg L <sup>-1</sup> , Mg <sup>2+</sup> : 16 mg L <sup>-1</sup> , NH <sub>4</sub> <sup>+</sup> : 502 mg L <sup>-1</sup> , suspended solid: 816 mg L <sup>-1</sup> , pH: 7.83, volume: 5 L	1 to 4 A	<ul><li>Applied current</li><li>Polarity inversion duration</li></ul>	[126]
Phosphorus recovery Polarity reversal	Electrochemical batch reactor, Anode = Cathode = Pt-coated Ti 4 pairs of electrodes were used Effective surface area: $126 \text{ cm}^2$ $d_{\text{elec}}$ : 1, 5 or 10 mm Polarity inversion period: 3 min	Real anaerobic digestion effluent pretreated with cationic organic polymer coagulant and centrifugally dehydrated  Total phosphorus: 83 mg L <sup>-1</sup> , dissolved phosphorus: 75 mg L <sup>-1</sup> , PO <sub>4</sub> <sup>3-</sup> : 75 mg L <sup>-1</sup> , Ca <sup>2+</sup> : 42 mg L <sup>-1</sup> , Mg <sup>2+</sup> : 20 mg L <sup>-1</sup> , NH <sub>4</sub> <sup>+</sup> : 443 mg L <sup>-1</sup> , suspended solid: 196 mg L <sup>-1</sup> , K <sup>+</sup> : 48 mg L-1, Na <sup>+</sup> : 66 mg L <sup>-1</sup> , Fe <sup>2+</sup> : 0.08 mg L <sup>-1</sup> , pH: 7.69, 8.0 or 8.4.	4 A (32 mA cm <sup>-2</sup> )	Interelectrode distance	[117]

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Process	Electrochemical system	Effluent / electrolyte	$j_{ m app}$ or $E$	Parameters	Ref
Water softening	Flow plant with undivided cell, Anode: $IrO_2/Ti$ DSA (100 cm <sup>2</sup> ) Cathode: 2D Ti mesh or 3D stainless steel mesh Geometric cathode surface area: $100 \text{ cm}^2$ (taken for current density calculation) $d_{\text{elec}}: 0.8 \text{ cm (flow-by)}$ $d_{\text{elec}}: 0.3 \text{ cm (flow-through)}$ Configuration flow-by vs. flow-through was compared	Simulated brackish water from electrodialysis concentrate,  Ca <sup>2+</sup> : 600 mg L <sup>-1</sup> , Mg <sup>2+</sup> : 200 mg L <sup>-1</sup> , HCO <sub>3</sub> <sup>-</sup> :  850 mg L <sup>-1</sup> , Cl <sup>-</sup> : 1000 mg L <sup>-1</sup> , NO <sub>3</sub> <sup>-</sup> : 800 mg  L <sup>-1</sup> , SO <sub>4</sub> <sup>2-</sup> : 770 mg L <sup>-1</sup> , Na <sup>+</sup> : 400 mg L <sup>-1</sup> Flow rate: 0.003 – 0.045 L min <sup>-1</sup> HCO <sub>3</sub> <sup>-</sup> : 0 – 5500 mg L <sup>-1</sup> Water hardness: 300 – 2380 mg L <sup>-1</sup> CaCO <sub>3</sub> Sodium hexametaphosphate (HMP) (scale inhibitor): 0.2 mg L <sup>-1</sup>	2.5 to 20 mA cm <sup>-2</sup>	<ul> <li>Flow-through vs. flow-by mode</li> <li>Flow rate</li> <li>Applied current density</li> <li>Bicarbonates         <ul> <li>concentration and water hardness</li> </ul> </li> <li>Presence of inhibitor</li> <li>Coupling vs. cascade mode to test scale-up for productivity</li> </ul>	[118]

**Abbreviations:** DSA: dimensionally stable anode, MMO: mixed metal oxide, Sn: tin, Pt: platinum, Ti: titanium, Ir: iridium, Ru: ruthenium, COD: chemical oxygen demand, TN: total nitrogen,  $d_{\text{elec}}$ : interelectrode distance and  $j_{\text{app}}$ : applied current density.

#### II.3. Pharmaceuticals removal by EAOPs

Once the deposits/fouling dominantly cover the electrode surfaces, without proper intervention, the efficiency of EAOP decreases. The capacity of the EAOP to mineralize pollutant as well as micropollutant decreases. In this thesis, tylosin, an antibiotic for veterinary use, was chosen as model micropollutant to check on the influence of electro-precipitation towards its removal.

The occurrence of pharmaceutical micropollutants in the effluent of WWTPs has been raising concern amongst different stakeholders. More and more pharmaceutical substances traceable in the wastewater have been categorized as emerging pollutants due to their potential hazard not only to human but also to other living organisms. Their detection in aquatic environment varies with the geographical activities: < 1 µg L<sup>-1</sup> at the outlet of municipal WWTP [127, 128], 6 µg L<sup>-1</sup> in the waste stream of dairy industry and reaching 30 µg L<sup>-1</sup> at the outlet of hospital waste [127]. Their continuous emission at very low concentration leads to accumulation in soil and aquatic environment. Prolonged and persistent exposure of aquatic life to the pharmaceutical substances has caused the development of antibiotic-resistant bacteria [129] as well as endocrine disruption, genotoxicity and wildlife poisoning [130, 131]. Thus, the occurrence of pharmaceutics in soils and water needs to be regarded with serious note. That is the reason why pharmaceutical micropollutants have been the case study of various degradation techniques including EAOPs.

Several pharmaceutical groups have been subjected to EAOP treatment in literature and some of them are recorded in **Table II.7**. Among them are (1) sulfonamides (sulfanilamide [132, 133], sulfamethazine [134-137], sulfamethoxazole [138, 139] and sulfachloropyridazine [140]), (2) tetracyclines [141], (3) betalactams (amoxicillin [142] and penicillin G [143]), (4) amphenicols (chloramphenicol [144, 145]), (5) bacteriostatic drugs (trimethoprim [146]), (6) nitro-imidazoles (metronidazole [147]), (7) macrolides (tylosin [148]), (8) quinolones (ciprofloxacin [149]), (9) nonsteroidal anti-inflammatory drugs (NSAID) (diclofenac [150] and ibuprofen [151]), (10) analgesic drugs (paracetamol [11, 12, 152]), (11) betablockers (atenolol, metoprolol and propranolol [153]), (12) timoprazoles (omeprazole [154]), (13) histamines (ranitidine [155]) and (14) bronchodilators (salbutamol [149]).

These pharmaceutical micropollutants were treated by various EAOPs such as EO, EF, coupled EF-EO, photo electro-Fenton via ultraviolet-A irradiation (PEF-UVA) and solar photo electro-Fenton (SPEF). Most of the investigations were done in synthetic solutions, in which the pharmaceuticals were spiked at the intended concentration. Still, several authors have used real wastewater matrices to take into consideration the real matrix influence (Table II.7).

Treatment of tylosin compound by EAOP has scarcely been reported in literature. Tylosin is an antibiotic drug commonly used to treat bacterial infection in small domestic animals such as cats, dogs and ferrets as well as livestock animals like sheep, goats and cows. Tylosin degradation by other techniques for example electrocoagulation [156], ozonation [157], photolysis [158] and advanced oxidation process such as photocatalyst [159-161], O<sub>3</sub>/H<sub>2</sub>O<sub>2</sub> [162] and nanosilver catalyst/peroxydisulfate [163, 164] have been documented. However, only one work from Ferrag-Siagh et al. [148] was found to report on the application of EAOP using EF as pretreatment to increase the biodegradability of tylosin. The treatment was proven efficient. Moreover, EAOPs such as EO and EF are much easier to setup as they operate in mild conditions with small footprint, without the need for stringent supply of UV source, O<sub>3</sub> or external H<sub>2</sub>O<sub>2</sub>. Furthermore, when EAOP is carried out in microfluidic reactor setup, operational cost of the treatment is significantly advantageous as compared to other AOPs. Due to this reason, tylosin degradation and mineralization using EO using submillimetric reactor will be performed in this PhD thesis. Compared to the EF, no addition of Fe<sup>2+</sup> catalyst as well as pH acidification are required. Moreover, microfluidic feature would offer mass transfer intensification advantage to the process which would be beneficial to oxidize pollutant of very low concentration. Eventually, some more insight on EAOP to oxidize tylosin will be elucidated, particularly in the matrix of real wastewater effluent.

Table II.7. Literature review on the EAOP treatments of different pharmaceutical pollutants.

Group	Pollutant	EAOP	Electrochemical system	Electrolyte	$\dot{j}_{ m app}$	Parameters	Ref
Sulfonamide	Sulfanilamide	ЕО	Undivided and divided cylindrical cell, Anode: BDD (3 cm²), Cathode: Stainless steel (3 cm²), $d_{\rm elec}$ : 1 cm	Synthetic solution, Sulfanilamide: $239 - 2390$ mg $L^{-1}$ , Na <sub>2</sub> SO <sub>4</sub> : 7 g L <sup>-1</sup> (undivided) and 70 g L <sup>-1</sup> (divided cell), pH: 2 - 6, volume: 100 mL	33 to 150 mA cm <sup>-2</sup>	<ul><li>Bulk pH</li><li>Applied current density</li><li>Initial sulfanilamide concentration</li></ul>	[132]
Sulfonamide	Sulfanilamide	EF SPEF	Flow pilot plant with undivided cell coupled to solar photo reactor, Anode: Pt sheet (20 cm²), Cathode: Carbon-PTFE-air diffusion (20 cm²), $d_{\text{elec}}$ : 1.2 cm, flow rate: 3.33 L min <sup>-1</sup>	Synthetic solution, Sulfanilamide: 239 – 1195 mg L <sup>-1</sup> , Na <sub>2</sub> SO <sub>4</sub> : 7 g L <sup>-1</sup> , pH: 3, Fe <sup>2+</sup> : 14 – 279 mg L <sup>-1</sup> (catalyst), volume: 100 mL	50, 100 and 150 mA cm <sup>-2</sup>	<ul> <li>Fe<sup>2+</sup> concentration</li> <li>Applied current density</li> <li>Initial concentration</li> </ul>	[133]
Sulfonamide	Sulfamethazine	EF PEF-UVA	Undivided cylindrical glass cell, Anode: BDD (3 cm²), Cathode: Carbon-PTFE-air diffusion (3 cm²), d <sub>elec</sub> : 1 cm, air flow rate: 300 mL min⁻¹	Synthetic solution, Sulfamethazine: $193 - 1930$ mg L <sup>-1</sup> , Na <sub>2</sub> SO <sub>4</sub> : 7 g L <sup>-1</sup> , Fe <sup>2+</sup> : $11 - 84$ mg L <sup>-1</sup> (catalyst), pH: $2 - 6$ , volume: $100$ mL	33 to 100 mA cm <sup>-2</sup>	<ul> <li>Bulk pH</li> <li>Fe<sup>2+</sup> concentration</li> <li>Applied current density</li> <li>Initial sulfamethazine concentration</li> </ul>	[134]
Sulfonamide	Sulfamethazine	EO EF	Undivided cell, Anode: Pt (20 cm²), Cathode: Carbon felt (208 cm²)	Synthetic solution, Sulfamethazine: 25 – 198 mg L <sup>-1</sup> , Na <sub>2</sub> SO <sub>4</sub> : 7 g L <sup>-1</sup> , Fe <sup>2+</sup> : 2.8 – 56 mg L <sup>-1</sup> , pH: 3, temp.: 18 – 45 °C, flow rate: 2 L min <sup>-1</sup> , V: 1 L	0.24 to 2.9 mA cm <sup>-2</sup>	<ul> <li>Applied current density</li> <li>Fe<sup>2+</sup> concentration</li> <li>Initial sulfamethazine concentration</li> <li>Operating temperature</li> <li>Biodegradability test</li> </ul>	[135]
Sulfonamide	Sulfamethazine	EO EO EF	Undivided cell, Anode: Pt, BDD, IrO <sub>2</sub> - RuO <sub>2</sub> /Ti or graphite felt (24 cm <sup>2</sup> ), Cathode: Graphite felt	Synthetic solution, Sulfamethazine: 29 mg L <sup>-1</sup> , Na <sub>2</sub> SO <sub>4</sub> : 7 g L <sup>-1</sup> , Fe <sup>2+</sup> : 11 mg L <sup>-1</sup> , pH: 3, volume: 300 mL	2.1 to 42 mA cm <sup>-2</sup>	<ul><li>Anode material</li><li>Applied current density</li><li>Comparison of EAOPs</li></ul>	[136]

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Group	Pollutant	EAOP	Electrochemical system	Electrolyte	$m{j}_{ m app}$	Parameters	Ref
Sulfonamide	Sulfamethazine	EF	Undivided cell, Anode: Pt Cathode: Carbon felt, carbon sponges (with different pore sizes) and stainless steel All geometric surface area: 21 cm², delec: 2.5 cm	Synthetic solution, Sulfamethazine: 29 mg L <sup>-1</sup> , Na <sub>2</sub> SO <sub>4</sub> : 7 g L <sup>-1</sup> , Fe <sup>2+</sup> : 11 mg L <sup>-1</sup> , pH: 3, volume: 250 mL	50 to 500 mA	<ul><li>Applied current</li><li>Cathode material</li></ul>	[137]
Sulfonamide	Sulfa- methoxazole	EO EF	Undivided cell, Anode: Pt mesh (EO) and Pt mesh or BDD (EF) (25 cm²) Cathode: Carbon felt (70 cm²)	$Synthetic \ solution,$ $Sulfamethoxazole: \ 21-329$ $mg \ L^{-1}, \ Na_2SO_4: \ 7 \ g \ L^{-1}, \ Fe$ $catalyst: \ 11 \ mg \ L^{-1} \ Fe^{2+}, \ or \ 11$ $mg \ L^{-1} \ Fe^{2+} + 13 \ mg \ L^{-1} \ Cu^{2+}$ $or \ 13 \ mg \ L^{-1} \ Cu^{2+}, \ pH: \ 3,$ $volume: \ 220 \ mL$	30 to 450 mA	<ul> <li>Applied current</li> <li>Initial sulfamethoxazole concentration</li> <li>Type of Fenton catalyst Fe<sup>2+</sup> and/or Cu<sup>2+</sup></li> </ul>	[138]
Sulfonamide	Sulfa- methoxazole	EO EF PEF-UVA	Undivided cell, Anode: RuO <sub>2</sub> /Ti mesh (16 cm <sup>2</sup> ) Cathode: EO: RuO <sub>2</sub> /Ti (16 cm <sup>2</sup> ) Others: activated carbon fiber (ACF) felt (16 cm <sup>2</sup> )	$Synthetic \ solution,$ $Sulfamethoxazole: 50-300$ $mg\ L^{\text{-}1},\ Na_2SO_4: 7\ g\ L^{\text{-}1},\ Fe^{2+}:$ $28-112\ mg\ L^{\text{-}1},\ pH: \ 3,$ $volume: \ 125\ mL$	7.5 to 31 mA cm <sup>-2</sup>	<ul> <li>Comparison between EAOPs</li> <li>Applied current density</li> <li>Fe<sup>2+</sup> concentration</li> <li>Initial sulfamethoxazole concentration</li> </ul>	[139]
Sulfonamide	Sulfachloro- pyridazine	EF	Undivided cell, Anode: Pt mesh or BDD (25 cm²) Cathode: Carbon felt (70 cm²)	Synthetic solution, Sulfachloropyridazine: 25 – 60 mg L <sup>-1</sup> , Na <sub>2</sub> SO <sub>4</sub> : 7 g L <sup>-1</sup> , Fe <sup>2+</sup> : 11 mg L <sup>-1</sup> , pH: 3, volume: 220 mL	30 to 350 mA	Applied current     Anode material	[140]
Tetracycline	Tetracycline	ЕО	Undivided cell, Anode: IrO <sub>2</sub> -RuO <sub>2</sub> /Ti DSA, Cathode: Carbon felt (240 cm <sup>2</sup> )	Synthetic solution, Tetracycline: 100 – 300 mg L <sup>-1</sup> , Na <sub>2</sub> SO <sub>4</sub> : 14 g L <sup>-1</sup> , pH: 3 – 9, volume: 0.4 L	500 to 1500 mA	<ul><li>Applied current</li><li>Bulk pH</li><li>Initial tetracycline concentration</li></ul>	[141]

Group	Pollutant	EAOP	Electrochemical system	Electrolyte	$oldsymbol{j}_{ ext{app}}$	Parameters	Ref
Betalactam	Amoxicilin	ЕО	Undivided cell, Anode: Carbon felt, carbon fiber, graphite, Pt, PbO <sub>2</sub> , IrO <sub>2</sub> -RuO <sub>2</sub> /Ti DSA or BDD (24 cm <sup>2</sup> ), Cathode: Stainless steel (24 cm <sup>2</sup> ), delec: 3 cm	Synthetic solution, Amoxicilin: 36 mg L <sup>-1</sup> , Na <sub>2</sub> SO <sub>4</sub> : 7 g L <sup>-1</sup> , pH: 5.3, volume: 250 mL	2.1 to 42 mA cm <sup>-2</sup>	<ul><li>Anode material</li><li>Applied current density</li></ul>	[142]
Nonsteroidal anti- inflammatory drug	Diclofenac	ЕО	Undivided cell, Anode: Pt or BDD (3 cm²), Cathode: Stainless steel (3 cm²) $d_{\text{elec}}: 1 \text{ cm}$	Synthetic solution, Diclofenac: 175 mg L <sup>-1</sup> , Na <sub>2</sub> SO <sub>4</sub> : 7 g L <sup>-1</sup> KH <sub>2</sub> PO <sub>4</sub> : 7 g L <sup>-1</sup> pH: 6.5 or without adjustment, volume: 0.1 L	17 to 150 mA cm <sup>-2</sup> (50 to 450 mA)	<ul><li>Bulk pH</li><li>Applied current density</li></ul>	[150]
Nonsteroidal anti- inflammatory drug	Ibuprofen	EF PEF-UVA SPEF	Undivided cell, Anode: Pt or BDD (3 cm2), Cathode: Carbon-PTFE-air diffusion (3 cm <sup>2</sup> ) O <sub>2</sub> flow rate: 20 mL min <sup>-1</sup>	Synthetic solution, Ibuprofen: 41 mg L <sup>-1</sup> , Na <sub>2</sub> SO <sub>4</sub> : 7 g L <sup>-1</sup> , pH: 2 – 6, Fe <sup>2+</sup> : 5.6 – 112 mg L <sup>-1</sup> , volume: 100 mL	3.3 to 100 mA cm <sup>-2</sup>	<ul> <li>Comparison between EAOPs</li> <li>Applied current density</li> <li>Fe<sup>2+</sup> concentration</li> </ul>	[151]
Analgesic	Paracetamol	EF EF-EO	Undivided cylindrical cell, Anode: Ti <sub>4</sub> O <sub>7</sub> plate (EO) or Pt/Ti mesh (EF) (24 cm <sup>2</sup> ) Cathode: Carbon felt (126 cm <sup>2</sup> ) Continuous air bubble: 1 L min <sup>-1</sup>	•Real WWTP effluent with conventional activated sludge process •Real WWTP effluent with membrane bioreactor (MBR) •Real-source urine treated with lab-scale MBR •Synthetic solution in 7 g L <sup>-1</sup> Na <sub>2</sub> SO <sub>4</sub> All matrices were spiked with 10 mg L <sup>-1</sup> paracetamol Fe <sup>2+</sup> : 11 mg L <sup>-1</sup> , pH: 3, volume: 250 mL	100, 250 and 300 mA	<ul> <li>Comparison between different background matrices of real wastewater effluent to mineralize acetaminophen</li> <li>Toxicity evaluation of different treated effluents</li> </ul>	[152]

Group	Pollutant	EAOP	Electrochemical system	Electrolyte	$\dot{J}$ арр	Parameters	Ref
Analgesic	Paracetamol	ЕО	Flow plant with undivided flow cell, Anode: BDD (50 cm²), Cathode: Carbon felt (50 cm²) $d_{\text{elec}}: 50 - 1000  \mu\text{m}$	•Synthetic solution in 0.14, 0.57 and 1.42 g L <sup>-1</sup> Na <sub>2</sub> SO <sub>4</sub> Volume: 200 mL •Real effluent of municipal WWTP (Reims, France) Volume: 500 mL	2 to 12 mA cm <sup>-2</sup>	<ul> <li>Applied current density</li> <li>Interelectrode distance</li> <li>Electrolytic conductivity</li> <li>Synthetic vs. real wastewater effluent</li> <li>Bulk pH</li> </ul>	[11]
				Both effluents were spiked with 15 mg L <sup>-1</sup> paracetamol, pH: 3 or neutral, flow rate:  0.43 L min <sup>-1</sup>			
Analgesic	Paracetamol	ЕО	Electro-mixing reactor, Anode: perforated BDD (14 cm <sup>2</sup> ), Cathode: Carbon felt (14 cm <sup>2</sup> ), $d_{\text{elec}}$ : 500 $\mu$ m	Synthetic solution, Paracetamol: 15 mg L <sup>-1</sup> , Na <sub>2</sub> SO <sub>4</sub> : 570 mg L <sup>-1</sup> , pH: neutral volume: 4 L	4 mA cm <sup>-2</sup>	<ul><li>Proof of concept</li><li>Electro-mixing vs. conventional flow-by reactor</li></ul>	[12]
Beta blocker	Atenolol, metoprolol and propranolol	EF SPEF	Recirculated flow electrochemical cell coupled to solar photo reactor, Anode: Pt or BDD (90.3 cm²), Cathode: Carbon- PTFE-air diffusion or carbon felt (90.3 cm²) 1 or 2 pairs of electrodes were tested, delec: 1.2 cm	Synthetic solution, Atenolol: 158 mg L <sup>-1</sup> , Metoprolol: 66 mg L <sup>-1</sup> Propranolol: 135 mg L <sup>-1</sup> Na <sub>2</sub> SO <sub>4</sub> : 14 g L <sup>-1</sup> , Fe <sup>2+</sup> : 28 mg L <sup>-1</sup> , temperature: 35 °C, pH: 3, flow rate: 4.2 L min <sup>-1</sup> , volume: 10 L	Carbon- PTFE-AD: 17 to 55 mA cm <sup>-2</sup> Carbon felt: 4.4 mA cm <sup>-2</sup>	<ul><li>Number of pairs of electrodes setting</li><li>EF vs. SPEF</li></ul>	[153]
Amphenicol	Chlor- amphenicol	ЕО	Flow plant with undivided cell  Anode: BDD or RuO <sub>2</sub> based MMO (78 cm <sup>2</sup> ), Cathode: Stainless steel (78 cm <sup>2</sup> ), $d_{\text{elec}}$ : 9 mm	Simulated urine waste of hospital effluent spiked with 100 mg L <sup>-1</sup> chloramphenicol	1.25 to 5 mA cm <sup>-2</sup>	<ul><li>Applied current density</li><li>Anode material</li></ul>	[144]

Chapter II

Group	Pollutant	EAOP	Electrochemical system	Electrolyte	$m{j}_{ m app}$	Parameters	Ref
Amphenicol	Chlor- amphenicol	EO EF PEF-UVA SPEF	Bench scale Undivided cylindrical cell Anode: Pt or BDD (3 cm <sup>2</sup> ) Cathode: Carbon-PTFE-air diffusion (3 cm <sup>2</sup> ), $d_{\text{elec}}$ : 1 cm Pilot scale Flow plant with undivided cell coupled to photo reactor Anode: Pt (90.2 cm <sup>2</sup> ) Cathode: Carbon-PTFE-air diffusion (90.2 cm <sup>2</sup> ) $d_{\text{elec}}$ : 1.2 cm	Synthetic solution, Chloramphenicol: 25 – 245 mg L <sup>-1</sup> Na <sub>2</sub> SO <sub>4</sub> : 7 g L <sup>-1</sup> Temperature: 35 °C pH: 3, Fe <sup>2+</sup> : 28 mg L <sup>-1</sup> Bench scale Volume: 100 mL, flow: magnetic stirring (800 rpm), air flow: 0.3 L min <sup>-1</sup> Pilot scale  Volume: 10 L, flow rate: 3.3 L min <sup>-1</sup> , air flow: 4.5 L min <sup>-1</sup>	33 to 100 mA cm <sup>-2</sup>	<ul> <li>Comparison between EAOPs</li> <li>Applied current density</li> <li>Initial chloramphenicol concentration</li> <li>Bench scale vs. pilot scale up scale</li> </ul>	[145]
Bacterio- static	Trimethoprim	EO EF PEF-UVA SPEF	Flow plant consisted of undivided electrochemical cell coupled to compound parabolic collectors and solar photo reactor  Anode: Pt or BDD ( $10 \text{ cm}^2$ )  Cathode: Carbon-PTFE-air diffusion ( $10 \text{ cm}^2$ ), $d_{\text{elec}}$ : 5 mm	Volume: 1.25 L Flow rate: 0.67 L min <sup>-1</sup> • Synthetic solution, Na <sub>2</sub> SO <sub>4</sub> : 7 g L <sup>-1</sup> , Fe <sup>2+</sup> : 2 – 8 mg L <sup>-1</sup> , pH: 3 – 4.5 • Real wastewater effluent of urban WWTP Both effluents were spiked with 20 mg L <sup>-1</sup> trimethoprim	2.5 to 150 mA cm <sup>-2</sup>	<ul> <li>Comparison of different EAOPs</li> <li>Fe<sup>2+</sup> concentration</li> <li>Applied current density</li> <li>Bulk pH</li> <li>Synthetic vs. real effluent</li> </ul>	[146]
Nitro- imidazole	Metronidazole	EF	Undivided cell, Anode: Pt (32 cm <sup>2</sup> ), Cathode: Graphite felt	Synthetic solution,  Metronidazole: 100 – 500 mg  L <sup>-1</sup> , Na <sub>2</sub> SO <sub>4</sub> : 7 g L <sup>-1</sup> , Fe <sup>2+</sup> : 2.8  – 56 mg L <sup>-1</sup> , pH: 2 – 9,  volume: 250 mL	0.04 to 0.45 mA cm <sup>-2</sup>	<ul> <li>Applied current density</li> <li>Bulk pH</li> <li>Initial metronidazole and Fe<sup>2+</sup>concentration</li> </ul>	[147]

Chapter II

Pollutant	<b>EAOP</b>	Electrochemical system	Electrolyte	$m{j}_{ ext{app}}$	<b>Parameters</b>	Ref
Omeprazole	ЕО	Undivided cell, Anode: Pt or BDD (3 cm²), Cathode: Carbon-PTFE-air diffusion (3 cm²), air flow rate: 300 mL min¹ $d_{\text{elec}}: 1 \text{ cm}$	Synthetic solution, Omeprazole: 17 – 169 mg L <sup>-1</sup> , NaH <sub>2</sub> PO <sub>4</sub> : 18 mg L <sup>-1</sup> H <sub>3</sub> PO <sub>4</sub> : 2.4 mg L <sup>-1</sup> , pH: 7, temperature: 35 °C, volume: 100 mL	33 to 150 mA cm <sup>-2</sup>	<ul><li>Anode material</li><li>Applied current density</li><li>Initial omeprazole concentration</li></ul>	[154]
Ranitidine	EO EF	Undivided cell, Anode: Pt (42 cm²), BDD (25 cm²) or IrO <sub>2</sub> -RuO <sub>2</sub> /Ti DSA (25 cm²) Cathode: Carbon felt (60 cm²)	Synthetic solution, Ranitidine: 31 mg L <sup>-1</sup> , Na <sub>2</sub> SO <sub>4</sub> : 7 g L <sup>-1</sup> , Fe <sup>2+</sup> : 6 mg L <sup>-1</sup> , pH: 3, temperature: 20 °C, volume: 230 mL	100 to 500 mA	<ul><li>Applied current</li><li>Comparison between EAOPs</li></ul>	[155]
Tylosin	EF	Undivided cell, Anode: Cylindrical Pt Cathode: Carbon felt (194 cm²)	Synthetic solution, Tylosin: 100 mg L <sup>-1</sup> , Na <sub>2</sub> SO <sub>4</sub> : 7.1 g L <sup>-1</sup> , Fe <sup>2+</sup> : 6 mg L <sup>-1</sup> , pH: 3, V: 0.6 L	300 mA	-	[148]
Ciprofloxacin  Sulfa- methoxazole  Salbutamol	ЕО	Flow plant with undivided cell Anode: BDD (69 cm²) Cathode: Zirconium (69 m²)	Synthetic solution, Ciprofloxacin: 14 – 25 mg L <sup>-1</sup> Sulfamethoxazole: 15 mg L <sup>-1</sup> Salbutamol: 16 mg L <sup>-1</sup> Urea: 5 – 38 mg L <sup>-1</sup> K <sub>2</sub> SO <sub>4</sub> : 17 g L <sup>-1</sup> (pH: 4) K <sub>2</sub> SO <sub>4</sub> : 3.5 g L <sup>-1</sup> (pH: 6)	1.4 to 7.2 mA cm <sup>-2</sup>	<ul> <li>Applied current density</li> <li>Electroactivity of compound to be treated</li> <li>Bulk pH</li> <li>Urea concentration</li> <li>Initial pollutant concentration</li> </ul>	[149]
	Omeprazole  Ranitidine  Tylosin  Ciprofloxacin  Sulfamethoxazole	Omeprazole EO  Ranitidine EF  Tylosin EF  Ciprofloxacin Sulfamethoxazole EO Salbutamol	$\begin{tabular}{lll} Undivided cell, \\ Anode: Pt or BDD (3 cm^2), \\ Cathode: Carbon-PTFE-air \\ diffusion (3 cm^2), air flow \\ rate: 300 mL min^{-1} \\ $d_{elec}$: 1 cm \\ \end{tabular}$ $\begin{tabular}{lll} EO & & & & & & & & \\ Ranitidine & & & & & & & \\ EO & & & & & & & \\ EO & & & & & & & \\ EO & & & & & & \\ EF & & & & & & \\ EO & & & & & & \\ EF & & & & & & \\ EO & & & & & \\ EF & & & & & & \\ EO & & & & & \\ EF & & & & & \\ EO & & & & & \\ EF & & & & & \\ EO & & & \\ EO & & & & \\ EO & & \\ E$	$\begin{tabular}{ c c c c c c c c c c c c c c c c c c c$	$ \text{Domeprazole } \\ \text{Domeprazole } \\ \text{EO} \\ & \begin{array}{c} \text{Undivided cell,} \\ \text{Anode: Pt or BDD (3 cm}^2), \\ \text{Cathode: Carbon-PTFE-air diffusion (3 cm}^2), air flow \\ \text{rate: } 300 \text{ mL min}^{-1} \\ \text{delec: 1 cm} \\ \text{Undivided cell,} \\ \text{Anode: Pt (42 cm}^2), BDD \\ \text{(25 cm}^2) \text{ or IrO}_2\text{-RuO}_2\text{/Ti} \\ \text{DSA (25 cm}^2) \\ \text{Cathode: Carbon felt (60 cm}^2) \\ \text{Cathode: Carbon felt (194 cm}^2) \\ \text{Cathode: BDD (69 cm}^2) \\ \text{Cathode: BDD (69 cm}^2) \\ \text{Cathode: Zirconium (69 m}^2) \\$	$ \text{Domeprazole } \\ \text{Domeprazole } \\ \text{Domeprazole } \\ \text{EO } \\ \begin{array}{c} \text{Undivided cell,} \\ \text{Anode: Pt or BDD (3 cm}^2), \\ \text{Cathode: Carbon-PTFE-air diffusion (3 cm}^2), air flow rate: 300 mL min^{-1} \\ delse: 1 cm \\ delse: 1 cm \\ \hline \\ \text{Undivided cell,} \\ \text{Anode: Pt (42 cm}^2), BDD \\ (25 cm}^2) \text{ or IrO}_2\text{-RuO}_2\text{-Ti DSA (25 cm}^2) \\ \text{Cathode: Carbon felt (60 cm}^2) \\ \hline \\ \text{Cathode: Carbon felt (60 cm}^2) \\ \hline \\ \text{Tylosin } \\ \text{EF } \\ \text{EF } \\ \begin{array}{c} \text{EO } \\ \text{Cathode: Carbon felt (60 cm}^2) \\ \text{Cathode: Carbon felt (194 cm}^2), BDD \\ \text{Cathode: Carbon felt (194 cm}^2) \\ \text{Cathode: BDD (69 cm}^2) \\ \text{Cathode: BDD (69 cm}^2) \\ \text{Cathode: Zirconium (69 m}^2) \\ \text{Cathode: Zirconium (69 m}^2)$

<u>Abbreviations</u>: EO: Electro-oxidation, EF: electro-Fenton, PEF: photo electro-Fenton, SPEF: solar photo electro-Fenton, UVA: ultraviolet A, BDD: boron-doped diamond, DSA: dimensionally stable anode, PTFE: polytetrafluoroethylene,  $j_{app}$ : applied current density, MMO: mixed metal oxide, Pt: platinum, Ti: titanium, Ir: iridium, Ru: ruthenium,  $d_{elec}$ : interelectrode distance and WWTP: wastewater treatment plant.

### II.4. Concluding remarks

Listed below are the critical subjects of experimentation, discussion and interpretation throughout this thesis:

- No mass transfer correlation in submillimetric parallel-plate electrochemical reactors has yet been established in literature either in laminar or turbulent regime.
- In a classical EAOP, mineral electro-precipitation is occuring on cathode. Under continuous operation without maintenance, the electrode surface will be totally passivated.
- No publication yet exists pertaining the formation of mineral scaling inside a microfluidic electrochemical reactor.
- The kinetic model to predict mineral electro-precipitation on cathode is scarcely available.
- Important parameters on mineral scaling for example interelectrode gap and electrode materials ought to be investigated in further details.
- The use of microfluidic reactors to treat wastewater merits more attention from scientific communities.
- No report has yet been made on the simultaneous fate of chlorinated and nitrogenous byproducts during wastewater treatment in submillimetric condition.

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### III. Material and methods

In this chapter, technical details on the chemical reagents used, the preparation of simulated effluents, the experimental pilot and electrochemical reactor setup are discussed. Moreover, the pre- and post-treatment procedures, analytical methods as well as the utilization of software are exposed.

### III.1. Chemical reagents

### **III.1.1.** Preparation of effluents

Calcium sulfate dihydrate (>90%), magnesium sulfate heptahydrate (>98%), sodium hydrogenocarbonate (>99.5%), sodium sulfate (>97%), sodium chloride (>99%) and potassium chloride (>99%) were purchased from VWR International (Fontenay-sous-Bois, France). Tylosin tartrate T6134 and sulfuric acid (95-97%) were obtained from Sigma Aldrich (Saint-Quentin-Fallavier, France). Ammonium sulfate (>99.5%) and ammonium sulfate (>99.5%) were supplied by Fluka Honeywell (Fontenay-sous-Bois, France). Peptone P and meat extract (lab-lemco) powders were provided from Thermo Fisher Scientific (Illkirch, France). These chemical reagents were used as purchased without further treatment.

### III.1.2. Analysis

Acetonitrile of high performance liquid chromatography – mass spectroscopy (HPLC LC-MS) grade (>99.9%) was purchased from VWR International (Fontenay-sous-Bois, France). Formic acid (>98%), titanium (IV) oxysulfate-sulfuric acid (27-31%) solution and H<sub>2</sub>O<sub>2</sub> standard solution (30% (w/w)) were obtained from Sigma Aldrich (Saint-Quentin-Fallavier, France). Nessler reagent, mineral stabilizer and polyvinyl alcohol dispersing agent were provided by HACH (Düsseldorf, Germany). Multi-element standard solution containing calcium (Ca), potassium (K), magnesium (Mg) and sodium (Na) of 1000 μg mL<sup>-1</sup> each was supplied by SCP Science (Courtabœuf, France). TIC, TOC and TN 1000 μg mL<sup>-1</sup> standard solutions were purchased from TechLab (Metz, France). Nitrate (NO<sub>3</sub>-), sulfate (SO<sub>4</sub><sup>2-</sup>) and chloride (Cl<sup>-</sup>) 1000 mg L<sup>-1</sup> standard solutions were also provided by Techlab (Metz, France). Nitrite (NO<sub>2</sub>-), chlorite (ClO<sub>2</sub>-), chlorate (ClO<sub>3</sub>-) and perchlorate (ClO<sub>4</sub>-) 1000 mg L<sup>-1</sup> standard solutions were purchased from Sigma Aldrich (Saint-Quentin-Fallavier, France).

#### III.1.3. Electrochemical cell characterization

Potassium hexacyanoferrate (II) trihydrate (98.5-102.0%), potassium hexacyanoferrate (III) (>99%) and sodium carbonate (>99.5%) were purchased from Sigma Aldrich (Saint-Quentin-Fallavier, France).

#### III.2. Preparation of effluent

#### III.2.1. Simulated effluents

Ion concentrations of simulated effluents were characteristic of reclaimed wastewater (RW) properties (Table III.1). The average pH of simulated effluents was adjusted to ~7.6. They all have an ionic strength of 0.02 mol L<sup>-1</sup>. In 2 L of ultrapure water (PureLab ELGA Classic, Veolia Water, Antony, France), basic effluent (BE) was prepared by dissolving 0.1014 g MgSO<sub>4</sub>.7H<sub>2</sub>O, 1.2887 g CaSO<sub>4</sub>.2H<sub>2</sub>O and 0.8400 g NaHCO<sub>3</sub>. 650 μL of H<sub>2</sub>SO<sub>4</sub> 1 M was added to adjust the pH. BE effluent without carbonates was prepared following the similar procedure by replacing NaHCO<sub>3</sub> with 0.4000 g Na<sub>2</sub>SO<sub>4</sub>. Intermediate effluent (IE) was prepared by dissolving 0.1014 g MgSO<sub>4</sub>.7H<sub>2</sub>O, 1.2887 g CaSO<sub>4</sub>.2H<sub>2</sub>O, 0.8400 g NaHCO<sub>3</sub>, 0.0132 g (NH<sub>4</sub>)<sub>2</sub>SO<sub>4</sub> and 0.0062 g NaH<sub>2</sub>PO<sub>4</sub>.2H<sub>2</sub>O in 2 L ultrapure water. 650 μL of H<sub>2</sub>SO<sub>4</sub> 1 M was added to adjust the pH. In 2 L ultrapure water, saline effluent (SE) was prepared by dissolving 1.2887 g CaSO<sub>4</sub>.2H<sub>2</sub>O, 0.0896 g MgSO<sub>4</sub>.7H<sub>2</sub>O, 0.8400 g NaHCO<sub>3</sub>, 0.0124 g Mg(NO<sub>3</sub>)<sub>2</sub>.6H<sub>2</sub>O, 0.0132 g (NH<sub>4</sub>)<sub>2</sub>SO<sub>4</sub>, 0.0062 g NaH<sub>2</sub>PO<sub>4</sub>.2H<sub>2</sub>O, 0.3810 g NaCl, 0.0650 g KCl and 800 μL of NO<sub>2</sub><sup>-</sup> 1000 μg mL<sup>-1</sup> standard solution. 650 μL of H<sub>2</sub>SO<sub>4</sub> 1 M was added to adjust the pH. Simulated wastewater (SW) consisted of similar salts composition as SE only with the addition of 32 mg L<sup>-1</sup> peptone and 22 mg L<sup>-1</sup> meat extract. 20 mg of tylosin tartrate was added in 2 L of effluent (10 mg L<sup>-1</sup>) in case where simulated effluent was spiked with tylosin.

#### III.2.2. Reclaimed wastewater effluent (RW)

RW was sampled at the outlet of Reims urban WWTP receiving a mixture of domestic and industrial waste streams. The plant treats  $65,000 \,\mathrm{m}^3$  of wastewater daily and has a treatment capacity of 470,000 people equivalents. The effluent was stored at -4 °C before usage and it was filtered with standard filter paper (Dutscher, Brumath, France) ( $\sim 10 \,\mu\mathrm{m}$  pore diameter) prior to experimentations.

**Table III.1**. Physicochemical properties of the effluent of municipal WWTP (Reims, France).

рН	IC	/ μS cm <sup>-1</sup>	Ca <sup>2+</sup> / mg L <sup>-1</sup>	Mg <sup>2+</sup> / mg L <sup>-1</sup>	TIC / mg-C L <sup>-1</sup>	TOC / mg-C L <sup>-1</sup>	TN / mg- N L <sup>-1</sup>
$7.57 \pm 0$	.21 1,0	021 ± 74	133 ± 8	$4.2 \pm 0.5$	$68 \pm 4$	$4.5 \pm 0.5$	$2.1 \pm 0.5$
$\mathbf{NH4}^{+}$ /	$NO_3$ -/	$NO_2^-$	PO <sub>4</sub> <sup>3-</sup> /	Cl <sup>-</sup> /mg	SO <sub>4</sub> <sup>2-</sup> /	<b>K</b> <sup>+</sup> / <b>mg</b>	Na <sup>+</sup> / mg
NH <sub>4</sub> <sup>+</sup> / mg L <sup>-1</sup>	NO <sub>3</sub> -/ mg L-1	NO <sub>2</sub> -/ mg L-1	PO <sub>4</sub> <sup>3-</sup> / mg L <sup>-1</sup>	Cl <sup>-</sup> / mg L <sup>-1</sup>	SO <sub>4</sub> <sup>2-</sup> / mg L <sup>-1</sup>	$ m K^+/mg$ $ m L^{-1}$	Na <sup>+</sup> / mg L <sup>-1</sup>

#### III.2.3. Electrolyte for electrochemical reactor characterizations

Mass transfer characterization of the electrochemical reactors as well as the determination of electroactive surface of electrodes were performed using electrochemical methods. They involved the use of ferro-ferricyanide solution prepared as follow: 84.4776 g of K<sub>4</sub>Fe(CN)<sub>6</sub>, 32.9240 g of K<sub>3</sub>Fe(CN)<sub>6</sub> and 105.99 g of Na<sub>2</sub>CO<sub>3</sub> were dissolved in 2 L of ultrapure water. It gave out a solution containing 0.05 mol L<sup>-1</sup> of K<sub>3</sub>Fe(CN)<sub>6</sub>, 0.1 mol L<sup>-1</sup> of K<sub>4</sub>Fe(CN)<sub>6</sub> and 0.5 mol L<sup>-1</sup> Na<sub>2</sub>CO<sub>3</sub> buffer.

#### III.3. Experimental setup

#### III.3.1. Electrochemical system

0.5 L of effluent was used in every experiment. The effluent was stored in a double-jacket stainless steel reservoir connected to a thermostatic bath (Bioblock Scientific, PolyScience, Niles, IL, USA) which can regulate the working temperature of experiments. A peristaltic pump (Masterflex, Cole-Parmer, Vernon Hills, IL, USA) was used to circulate the effluent at the desired flow rate (10 to 800 mL min<sup>-1</sup>) in batch recirculation mode. The electrochemical setup consisted of a parallel-plate reactor (Fig. III.1). Its body was made of poly (methyl methacrylate) (PMMA), a material chemically and thermally resistant in the applied operating conditions. Rubber seals were designed, built-in, to provide tightness and avoid leakage. Both anode and cathode were planar rectangles, facing one another. BDD (DiaCCon, Fürth, Germany), 2 mm in thickness, was coated on both sides with a Niobium substrate by 12 µm. BDD was always employed as anode. Pure platinum plate (Ögussa, Vienna, Austria) was adopted as anode in some preliminary studies. 316L stainless steel (Gantois Industries, Saint-Dié-des-Vosges, France), graphite plate (Final Advanced Materials, Didenheim, France) and carbon paper (Ion Power, München, Germany) were used as cathode. The width (W) and length (L) of the planar electrodes in contact with the electrolyte were 5 and 10 cm respectively giving out a geometric surface area of 50 cm<sup>2</sup>. A spacer made of polytetrafluoroethylene (PTFE) with varying thicknesses of 100, 250, 500, 1000, 1500, 2000 and 3000 µm (Bohlender, Grünsfeld, Germany) was used to separate the cathode and anode. Thus, they defined the varying  $d_{\text{elec}}$ investigated in this work, distinguished from micrometric progressively into a typical millimetric

#### Chapter III

configuration. A HMP4040 model current generator (Rohde & Schwarz, Meudon-la-Forêt, France) was used to run the electrochemical reactor under galvanostatic mode. A range of 20 to 800 mA applied current (corresponding to 0.4 to  $16 \, \text{mA cm}^{-2}$  current density) was tested throughout the study. For the experiments using a potentiostat (Ametek, Massy, France), the electrochemical cell was connected to the potentiostat under a 3-electrode configuration. Silver-saturated silver chloride (AgAgCl) was used as reference electrode (RE). It was inserted into the cell at the bottom part, near the electrolyte inlet stream as depicted in Fig. III.2. The electrolyte was introduced by forced convection from the bottom of the cell flowing upward. The positioning of the RE (lower part of the cell) took advantage of the gravitational force to ensure stable measurement conditions by avoiding: firstly, engorgement in the case of unstable effluent flow and secondly, perturbation due to gas evolution. Furthermore, the Ag-AgCl reference electrode was positioned very close to the working electrode (500  $\pm$  100  $\mu$ m) in order to avoid ohmic loss in solution.

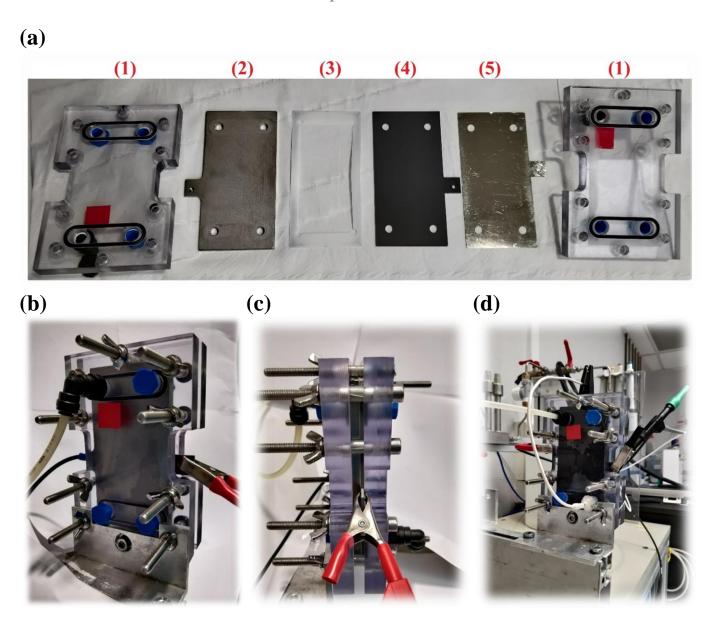
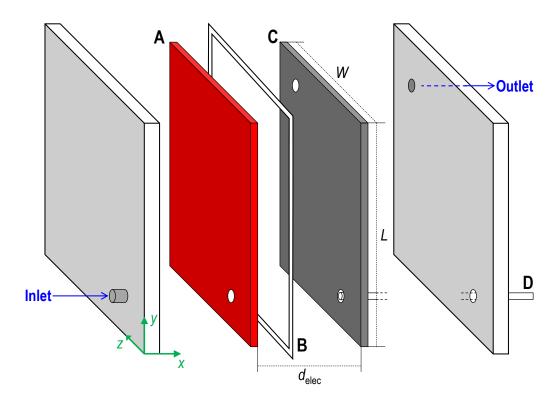


Fig. III.1. (a) Disassembled parallel-plate cell with 1: PMMA cell plate, 2: stainless steel cathode, 3: PTFE spacer, 4: BDD plate anode or 5: Pt plate anode, (b) side view of assembled electrochemical cell working in 2-electrode configuration, (c) front view of thin-film electrochemical reactor working in 2-electrode configuration and (d) side view of the cell working under 3-electrode configuration.



**Fig. III.2**. An exploded view of the parallel-plate electrochemical reactor. A: anode or counter electrode, B: PTFE spacer acting as interelectrode separator, C: cathode or working electrode, D: Ag-AgCl reference electrode,  $d_{\text{elec}}$ : interelectrode distance, W: width and L: length of electrodes.

#### III.3.2. Pretreatment of electrodes

Prior to utilization, stainless steel and graphite plates were preconditioned chemically by soaking them in 300 mL of sulfuric acid solution at 0.25 M overnight to get rid of possible remaining electroprecipitate. Additionally, the graphite was pretreated electrochemically to remove plausible deposition residue in the pores. 0.25 M of sulfuric acid was used as electrolyte, the graphite plate was positioned as anode, whilst a carbon felt (Mersen, Gennevilliers, France) served as cathode. The graphite anode was oxidized at 300 mA using a current generator and the oxidation pretreatment lasted an hour. Meanwhile, the carbon paper was used as received without pretreatment. Carbon paper was cut to take similar form as other cathodes as illustrated in **Fig. III.3**. Each carbon paper electrode was used only once, to ensure repeatability.

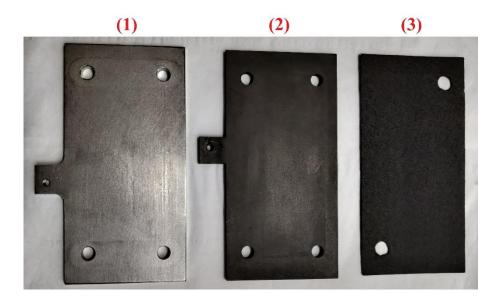


Fig. III.3. (1) Stainless steel, (2) graphite and (3) carbon paper cathodes.

#### III.3.3. Post treatment and recovery process

After each experiment, sulfuric acid solution of 0.25 M was used to thoroughly wash and recover precipitates occurring in different compartments of the experimental pilot. Anode was immersed in 50 mL of the acidic solution for 30 min. 300 mL of acidic solution was recirculated inside the experimental pilot plant during 1 h. Another 300 mL of 0.25 M sulfuric acid solution was used to immerse cathode overnight. This step served as both post-treatment as well as pretreatment of the cathode regeneration for the next experiment. The purpose was to re-dissolve the mineral deposits back into ionic form by lowering the pH. For recovery process, the acid washing solution of different compartments were then filtered through Phenex<sup>TM</sup> regenerated cellulose 0.45 μm filter (Phenomenex, Le Pecq, France) and were analyzed by inductively coupled plasma optical emission spectroscopy (ICP-OES) to measure the Ca and Mg concentrations. Knowing different concentrations of cations in different compartments of experimental pilot, mass balance of cations was performed to observe the distribution of mineral precipitates formed in and out of the electrochemical reactor in addition to attesting a quasi-complete recovery of cations.

Mass balance and percentage recovery of i elements (Ca and Mg) are defined in **Eq.** (III.1) and **Eq.** (III.2):

$$m_{i,recovered} = m_{i,fin} + m_{i,anode} + m_{i,cathode} + m_{i,tube}$$
 (III.1)

$$(m_{i,recovered} / m_{i,init}) \times 100 = \% recovery$$
 (III.2)

where  $m_{i,recovered}$  is the total mass (in mg) of *i* recovered from different compartments,  $m_{i,init}$  the initial mass of *i* element in electrolyte before electrolysis,  $m_{i,fin}$  the mass of *i* that remains in the

electrolyte at the end of electrolysis,  $m_{i,anode}$  the mass of i recovered from the surface of anode,  $m_{i,cathode}$  the mass of i recovered from the surface of cathode and  $m_{i,tube}$  the mass of i recovered from the surface of pilot tubing.

#### III.4. Analytical methods

#### III.4.1. Inductively coupled plasma optical emission spectrometry (ICP-OES)

Ca and Mg elemental concentrations were monitored using ICP-OES Thermo iCAP 6000 (Thermo Fisher, Noisy-Le-Grand, France). Collected samples were filtered using Phenex<sup>TM</sup> 0.45 μm filter and were dissolved in 0.5 M H<sub>2</sub>SO<sub>4</sub> background matrix inside ICP tube with 10 mL capacity. Using ICP-OES, the samples were nebulized, vaporized, atomized, ionized and excited thermally by means of argon plasma and ceramic torch. A radio-frequency generator produced a frequency of 1,100 kHz to attain a temperature up to 10,000 K. They were then detected and quantified by optical emission spectrometer at wavelengths of 279.55 nm, 280.27 nm and 285.21 nm for Mg and 317.93 nm, 318.12 nm and 422.67 nm for Ca. The calibration was done using external calibration method. Standard solutions ranging from 0.5 to 5 mg L<sup>-1</sup> Mg as well as from 5 to 50 mg L<sup>-1</sup> Ca were prepared from the Ca, K, Mg and Na multi-element 1000 μg mL<sup>-1</sup> standard solution. Control solutions with known concentrations of Mg and Ca were analyzed every 10 analyses of samples to check for any drift in instrument measurement.

#### III.4.2. TOC/TN analysis

Total inorganic carbon (TIC), total organic carbon (TOC) and total nitrogen (TN) were measured using Shimadzu  $V_{CSH}$  TOC coupled to TNM-1 analyzer (Marne-La-Vallée, France). The analyzer was equipped with Shimadzu ASI-V autosampler (Marne-La-Vallée, France) that could accommodate 68 40 mL vials and automate analysis process. The samples were filtered and diluted accordingly before being stored in TOC glass vial (40 mL capacity). TIC was programmed to be analyzed separately with the TOC and TN. For the TIC analysis, a small amount of hydrochloric acid was added to acidify the sample (pH < 3). All carbonates in the sample were then converted to carbon dioxide (CO<sub>2</sub>). The acidified sample was sparged with carrier gas (purified air) to volatilize the converted  $CO_2$ . It was dehumidified in a dehumidifier before passing through halogen scrubber into a sample cell of a non-dispersive infrared detector (NDIR). The NDIR analog signal was converted to TIC concentration by intermediary of TIC calibration curve. A calibration curve was previously performed via an autodilution program of the  $V_{CSH}$  TOC instrument. A series of TIC standard solutions varying between 1, 2, 5, 10, 12.5, 25 and 50 mg-C  $L^{-1}$  were analyzed to establish the curve. Next, the measurement of TOC and TN were carried out simultaneously. Due to relatively elevated

inorganic carbon content in comparison to organic carbon in all investigated effluents throughout this thesis, TOC was analyzed following the non-purgeable organic carbon (NPOC) procedure. Furthermore, volatile organic species was not present in the synthetic effluent as well as in the WWTP effluent. Using this method, a small quantity of hydrochloric acid was added into the sample and it was sparged with purified air to chase out the CO<sub>2</sub>. As a result, all inorganic carbon content of the sample was eliminated. The remaining organic carbon was carried by the carrier gas (purified air) into a catalyst-filled combustion tube and heated to 720 °C. The sample was decomposed into CO<sub>2</sub>. The combustion products were cooled, dehumidified and passed through the halogen scrubber. The CO<sub>2</sub> was measured by NDIR and the signal was converted into NPOC concentration by means of NPOC calibration curve. A NPOC calibration curve was established between 0.5 and 10 mg-C L<sup>-1</sup> employing TOC commercial standard solution and using autodilution program. When the sample was introduced into the catalytic combustion chamber at 720 °C, if nitrogenous compounds were present, they were decomposed into nitrogen monoxide. The carrier gas then containing nitrogen monoxide was cooled and dehumidified before being sent into a chemiluminescence gas analyzer. The nitrogen monoxide detection signal was converted into TN quantification by intermediary of a TN calibration curve. It was previously established using TN standard solutions, whose concentrations varied between 1, 2, 5, 10, 16.67, 25 and 30 mg-N L<sup>-1</sup> autodiluted by the instrument.

#### III.4.3. Chromatography

#### III.4.3.1. Ionic chromatography

NO<sub>2</sub>-, NO<sub>3</sub>-, Cl<sup>-</sup>, ClO<sub>2</sub>-, ClO<sub>3</sub>-, ClO<sub>4</sub>- and SO<sub>4</sub><sup>2</sup>- anion concentrations were measured via ionic chromatography Dionex<sup>TM</sup> ICS-6000 Capillary HPIC<sup>TM</sup> (Thermo Scientific, Noisy-Le-Grand, France). Standard solutions containing each anion of known concentrations were prepared from their corresponding commercial standard solution of 1000 μg mL<sup>-1</sup>. The range of calibration for each anion is given as follow. NO<sub>2</sub>-, ClO<sub>2</sub>- and ClO<sub>3</sub>- were calibrated from 0.5 to 40 mg L<sup>-1</sup>, NO<sub>3</sub>- from 0.5 to 30 mg L<sup>-1</sup>, Cl<sup>-</sup> was calibrated from 1 to 300 mg L<sup>-1</sup> while SO<sub>4</sub><sup>2</sup>- from 1 to 480 mg L<sup>-1</sup>. AS19 IC Dionex<sup>TM</sup> IonPac<sup>TM</sup> anion-exchange column (25 cm × 4 mm) (Thermo Scientific, Noisy-Le-Grand, France) equipped with AG19 Dionex<sup>TM</sup> IonPac<sup>TM</sup> column guard (Thermo Scientific, Noisy-Le-Grand, France) was used. The pump worked in isocratic mode and the eluent flow rate was set at 1 mL min<sup>-1</sup>. The autosampler tray was regulated at 20 °C, the column of detector compartment was maintained at 30 °C and the suppressor of conductivity detector was regulated at 35 °C. All samples were filtered using Phenex<sup>TM</sup> 0.45 μm filter and kept inside HPLC 1.5 mL vial prior to analysis.

# III.4.3.2. High performance liquid chromatography – photo diode array detection (HPLC-PDA)

HPLC-PDA was employed to quantify the evolution of tylosin concentration during the EAOP treatment. The collected sample was filtered using Phenex<sup>TM</sup> 0.45 μm filter, undiluted and stored in HPLC 1.5 mL vial. Tylosin calibration curve was established using tylosin solution containing 0.5, 1, 2, 5, 7.5, 10 and 20 mg L<sup>-1</sup> tylosin. The HPLC chain consisted of pump unit, column oven, autosampler section and detection unit. Kinetex® reversed phase C18 column with 2.6 μm and 100 Å particle and pore sizes respectively, 4.6 mm in internal diameter and 100 mm in length (Phenomenex, Le Pecq, France) was the core of the separation technique. The pump worked in binary gradient mode pumping 35% of solvent A (ultrapure water plus 1% v/v formic acid) and 65% of solvent B (acetonitrile). The eluent flow rate was 0.6 mL min<sup>-1</sup>. The oven, in which the column was kept, was operated at 25 °C. A thermo-regulated rack was used for sampling. Needle rinse mode was programmed before and after aspiration and each dip lasted 3 s. Autosampler chamber was set at 20 °C during analysis and it switched to 4 °C when the analysis ended for proper samples storage. 50 μL of sample was injected for tylosin concentration analysis. The HPLC was coupled to a PDA detector. The detector cell unit was set at 30 °C. UV absorbances at the wavelengths ranging from 190 to 350 nm were registered. The maximal absorbance of tylosin molecule was recorded at 290 nm.

#### III.4.4. Spectrophotometry

Quantification of  $NH_4^+$ , active chlorine and  $PO_4^{3-}$  were performed using a HACH spectrophotometer (Lognes, France). The colorimetric quantification of  $NH_4^+$  was done with the Nessler method (Method 8038, 0.02 to 2.50 mg  $NH_3$ -N  $L^{-1}$ ) and that of free chlorine by DPD method (N,N diethyl-phenylenediamine) (Method 8021, 0.02 to 2.00 mg- $Cl_2$   $L^{-1}$ ). The PhosVer3 method (ascorbic acid) (Method 8048, 0.02 to 2.50 mg- $PO_4^{3-}$   $L^{-1}$ ) was used to dose  $PO_4^{3-}$ .

Hydrogen peroxide ( $H_2O_2$ ) concentration was measured via  $H_2O_2$ -Ti<sup>4+</sup> complexation method [1]. 100  $\mu$ L of titanium (IV) oxysulfate sulfuric acid solution was put inside a glass vial tube before adding 4 mL of sample to be analyzed. If  $H_2O_2$  was present, the mixture gave out a yellowish color. A UV-2600 spectrophotometer (Shimadzu, Marne-La-Vallée, France) was used to detect maximum absorbance of the complex at 410 nm wavelength.  $H_2O_2$  concentration was deduced according to the Beer-Lambert law ( $H_2O_2$  molar extinction coefficient of 43  $M^{-1}$  cm<sup>-1</sup>) from a calibration curve established using a series of  $H_2O_2$  standard solutions with concentrations varying from 0.01 to 1 mM.

Fluorescence spectrophotometry was performed to inspect the presence of dissolved organic matter in the tested effluent. Fluorescence spectrophotometer F-2500 (Hitachi, Düsseldorf, Germany) was

used to achieve the purpose. 5 mL of sample was filled inside quartz cuvette for the analysis. Wavelength scan operation, in synchronous scan mode between 230 to 600 nm at 300 nm min<sup>-1</sup>, was performed for each sample. The fluorescence intensity of organic content of the effluent in function of excitation wavelength was then captured.

# III.4.5. Scanning electronic microscopy (SEM) coupled to energy dispersive X-ray (EDX) analysis

Images of electro-precipitates on carbon paper cathode were taken using a SEM JEOL JSM-6490 (Croissy-sur-Seine, France) coupled with EDX analysis. The latter allowed determining the elemental analysis of electro-precipitates. The SEM scan was done on carbon paper because only this cathode material could be cut to follow the 3 cm diameter SEM metal sampler, without affecting the deposits. Prior to SEM scan, a cut carbon paper sample was fixed onto the metal sampler with the help of a double-sided carbon adhesive. The sample surface was then metalized by means of gold nanoparticles using JEOL JFC-1100 Ion Sputter (Croissy-sur-Seine, France). During SEM measurements, the electronic beam was set at 5 kV accelerator volt. The image magnification varied between 80× to 1500×. For EDX elemental analysis, the accelerator volt was set to 20 kV using a magnification factor of 400×.

#### III.4.6. Electrochemical methods

A potentiostat (Ametek, Massy, France) was equipped with a PMC-1000 channel. It could perform both direct current (DC) and alternative current (AC) experiments up to  $\pm 10$  V and  $\pm 2$  A respectively. Owing to the integration of frequency response analyzer (FRA) to the potentiostat, electrochemical impedance spectroscopy (EIS) could be carried out between 10  $\mu$ Hz to 1 MHz. It offered AC voltage amplitude between 0.1 to 1000 mV root mean square (RMS).

#### III.4.6.1. Chronoamperometry and chronopotentiometry

The PMC-1000 potentiostat allowed chronoamperometry and chronopotentiometry experiments by applying either fixed potential or fixed current, respectively. In either case, Ag-AgCl reference electrode was used. Chronoamperometry experiments were conducted by applying a constant potential of -1.6 V/Ag-AgCl under different electrochemical reactor setups. The role of the applied electrode potential was investigated during the study. Meanwhile, chronopotentiometry experiments were performed under multiple reactor configurations by varying pairs of electrodes as well as  $d_{\rm elec}$ . The purpose was to get the corresponding values of cathode and anode applied potentials. For that

purpose, a fixed current of either -200 or +200 mA was applied on the targeted working electrode under the tested cell configuration.

#### III.4.6.2. Cyclic voltammetry (CV) and linear scan voltammetry (LSV)

CV and LSV are powerful tools to study the electrochemical behavior of an electrolyte inside a specific electrochemical reactor. They involve a sweeping of a range of potential, at a certain scan rate and the intensity response of the electrochemical system is recorded. A voltammogram is the outcome of a CV or LSV measurement. Different ranges of potential sweeping as well as different scan rates were used depending on the objective of the study. In the mass transfer characterization study from submillimetric into millimetric electrochemical reactor, the potential was swept between 0.2 to -1.6 V/Ag-AgCl at a scan rate of 2.5 mV s<sup>-1</sup> [2, 3]. The electrochemical behavior of investigated effluents inside different electrochemical reactor configurations, LSV was performed sweeping a potential range of 0.5 to -1.8 V/Ag-AgCl at 10 mV s<sup>-1</sup> scan rate. Elsewhere, during the estimation of electroactive surface of electrodes, CV was carried out between 0.6 to -0.8 V/Ag-AgCl with using a scan rate of 10 mV s<sup>-1</sup> [4, 5].

#### III.4.6.3. Electrochemical impedance spectroscopy (EIS)

The EIS technique is based on small amplitude AC perturbations at different AC frequencies and the analysis of the system response gives out the information on electrochemical and physical parameters, reactions kinetic as well as the surface morphology at the electrode/electrolyte interface. In this thesis, EIS was conducted in galvanostatic mode covering a frequency range between 100 kHz to 100 mHz. The perturbation amplitude ( $\Delta I_{AC}$ ) varied between 5 and 10 mA RMS according to EIS spectra with 10 points per decade (ppd). The EIS analysis was programmed before electrolysis (t = 0) as well as at 0.5, 1, 2, 3, 4 and 5 h of galvanostatic electrolysis at an applied 200 mA (i.e. 4 mA cm<sup>-2</sup>) DC. The EIS parameters were extracted by means of equivalent electrical circuit (EEC) model using ZSimpwin® commercial software.

#### III.5. Modeling software

#### III.5.1. Kramers-Kronig test

Kramers-Kronig transform (KKT) method was used to check the applicability of EIS technique by comparing the experimental EIS plot with a mathematical-conformed spectrum. To run the KKT test, two software were used: (1) KKT program offered by the Zsimpwin® and (2) the KKT program developed by Dr. Bernard Boukamp from the University of Twente [6, 7]. If the two spectra coincided

within acceptable deviation margin (evaluated by the software in chi-squared residues ( $\chi^2$ )), the experimental EIS spectra were validated.

#### III.5.2. Zsimpwin®

EIS raw data were collected from VersaStudio® which was the interface for Ametek® potentiostat. Using Zsimpwin®, the EIS parameters were extracted from the raw data by intermediary of EEC model.

#### III.5.3. Aquasim©

Aquasim© provides an interface where chemical reactions, physicochemical equilibriums and constants, mathematical equations as well as kinetic laws of multiple species can be easily inserted and accounted for [8]. Furthermore, the software already comes with the embedded codes to visualize the plot of the resulting model curve and can be compared directly with the experimental one. Thus, Aquasim© was used to complement the mathematical models developed in this thesis. It was used to simulate the evolution of Ca<sup>2+</sup> and Mg<sup>2+</sup> concentrations during their participation in the electroprecipitating reactions.

#### III.6. Fitting evaluation criteria between experimental and modeling data

The fitting between experimental data and theoretical model was evaluated using either root mean square error (RMSE), model efficiency (ME) and/or index of agreement (IOA) given as Eq. (III.3), Eq. (III.4) and Eq. (III.5), respectively [9]:

$$RMSE = \sqrt{\frac{\sum_{i=1}^{K} (y_i - y_i')^2}{K}}$$
 (III.3)

$$ME = 1 - \frac{\sum_{i=1}^{K} (y_i - y_i')^2}{\sum_{i=1}^{K} (y_i - y_M)^2}$$
 (III.4)

$$IOA = 1 - \frac{\sum_{i=1}^{K} (y_i - y_i')^2}{\sum_{i=1}^{K} (|y_i' - y_M| + |y_i - y_M|)^2}$$
(III.5)

where K is the number of steps of checked values,  $y'_i$  is the experimental value,  $y_i$  is the corresponding simulated value and  $y_M$  is the average of simulated value.

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# IV. Mineral cathodic electro-precipitation and its kinetic modeling in thin-film microfluidic reactor during advanced electro-oxidation process

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#### IV.1. Introduction

The occurrence of electro-precipitation is a major drawback within the application of EAOPs and more generally electrolysis to treat wastewater. While treating water of varying concentrations of calcium (Ca<sup>2+</sup>) and magnesium (Mg<sup>2+</sup>) ions), a layer of insulating deposit can be formed on the surface of cathode. Despite low concentration of Ca<sup>2+</sup> and Mg<sup>2+</sup> in an effluent, in case of the operation of electrochemical cells in continuous-mode such, most of the time, encountered in industrial applications, the deposit progressively passivates the whole electrode surface. Sequencing polarity inversion might be applied as counter-measure but it might take a toll on treatment efficacy, duration of operation and overall operational cost. Furthermore, cathode in operation must be able to tolerate high anodic current during the inverted polarization step given the fact that some cathode materials have been reported to be corroded in such conditions [1]. Upon treating water with high Ca<sup>2+</sup> and Mg<sup>2+</sup> content, sequencing inversion of polarity might not be the best strategy since the passivation of electrode occurs anyhow. Eventually, without regular intervention, electrochemical cells become very inefficient or probably are no longer operational.

The mechanism of formation of deposits has been well established in the literature [2, 3]. When the electrode is polarized at a sufficiently negative potential, dissolved  $O_2$  is reduced to hydroxide ions  $(OH^-)$  according to Eq. (II.9) resulting in an increase of local pH [4, 5].

$$O_2 + 2H_2O + 4e^- \rightarrow 4OH^-$$
 (II.9)

At a more cathodic potential, reduction of solvent  $(H_2O)$  can take place according to **Eq.** (II.10), which also produces hydroxide ions as well as  $H_2$ .

$$2H_2O + 2e^- \rightarrow H_2 + 2OH^-$$
 (II.10)

As shown in **Eqs. (I.11)** and **(IV.1)**, in the presence of Mg<sup>2+</sup> and Ca<sup>2+</sup>, deposits of hydroxide can be produced. Moreover, the pH increment also shifts thermodynamic ratio of carbonates/bicarbonates (**Eq. (I.12)**) favoring the production of carbonate scales as written in **Eqs. (I.13)** and **(IV.2)**.

$$Mg^{2+} + 2OH^- \leftrightarrow Mg(OH)_2 \tag{II.11}$$

$$OH^- + HCO_3^- \leftrightarrow H_2O + CO_3^{2-}$$
 (II.12)

$$Ca^{2+} + CO_3^{2-} \leftrightarrow CaCO_3 \tag{II.13}$$

$$Ca^{2+} + 20H^- \leftrightarrow Ca(0H)_2 \tag{IV.1}$$

$$Mg^{2+} + CO_3^{2-} \leftrightarrow MgCO_3 \tag{IV.2}$$

Different studies have already demonstrated the formation of aforementioned deposits in electrochemical cell while operating with different types of water with varying contents of Ca<sup>2+</sup> and Mg<sup>2+</sup>. They are characterized by their source and application domain: synthetic water for cooling system in cooling towers and oil refineries [6, 7], artificial [8, 9] and natural groundwater [10, 11], mineral potable water [12, 13], artificial [3, 14-16] and real seawater [17, 18]. Interestingly, the reported works were all conducted on the classical rotating disk electrode (RDE) possessing effective exposed surface area of 0.2 cm<sup>2</sup> [3, 8-12, 14, 17], 0.5 cm<sup>2</sup> [6, 7, 19], 1 cm<sup>2</sup> [16, 20, 21] and 1.5 cm<sup>2</sup> [15]. Either carbon steel, gold or stainless steel was used as the working electrode playing the role of cathode. To the best of authors' knowledge, the study of electro-precipitation outside aforementioned configuration is scarce.

Due to the omnipresence of calcium and magnesium in water, further study to understand and more particularly to avoid the formation of insulating deposition at larger scale seems primordial. It is within this context that a detailed investigation has been carried out to understand the influence of electrolytic composition towards the formation of mineral deposits on electrode surface during EAOP. The objective is to establish operating conditions in-which scaling does not occur, or, is minimized owing to the benefits attributed by a microfluidic cell. Moreover, addition of chemical reagents is no longer compulsory in microreactor; hence, interference possibly originating from the supporting electrolyte is avoided. Magnesium hydroxide (Mg(OH)<sub>2</sub>), calcium carbonate (CaCO<sub>3</sub>) and magnesium carbonate (MgCO<sub>3</sub>) deposits are studied due to the ubiquity of Mg<sup>2+</sup>, Ca<sup>2+</sup> and HCO<sub>3</sub><sup>-</sup> /CO<sub>3</sub><sup>2-</sup> ions in water sources. Results presented in this chapter are the first to elucidate the phenomenon of passivation in microfluidic reactor, a configuration very distinctive to those reported in the literature where RDE were used as working electrode. The geometry and configuration of microfluidic cell adopted in this work are closely representative to reactors used in industries. Notably, the difference between electrode surface in this present study is 25 to 250 times bigger compared to RDEs (0.2 to 2 cm<sup>2</sup>) used in the literature. The present design is more adapted for further applications on a larger scale. Synthetic water is used in this work, but it closely mimics the average characteristic of effluent of municipal WWTPs particularly the concentration of Ca<sup>2+</sup>, Mg<sup>2+</sup> and carbonates as well as pH and ionic conductivity. Furthermore, it is the first time that the influence on electro-precipitation of hydroxyl radicals produced at high O<sub>2</sub> evolution overvoltage anode is studied. Novel theoretical models are developed in this paper where the objectives are two-fold; first, assumptions made to establish the models lead to a better understanding of the mechanism of mineral electro-precipitation if they occur. Second, these models are the first to be reported to predict the evolution of electro-precipitating ions, particularly under the configuration of thin-film microfluidic reactor.

#### IV.2. Experimental section

Parallel-plate electrochemical cell with BDD and stainless steel as anode and cathode respectively was used. It was configured under submillimetric condition by using 500  $\mu$ m  $d_{elec}$  between anode and cathode. The electrolyte was recirculated at 100 mL min<sup>-1</sup>. Effluent with and without carbonates contained sources of Ca<sup>2+</sup> (150 mg L<sup>-1</sup>) and Mg<sup>2+</sup> (5 mg L<sup>-1</sup>) dissolved in ultrapure water. Effluent with carbonates contained a concentration of 60 mg-C L<sup>-1</sup>. Initial values of pH and ionic conductivity were always adjusted to 7.6 and 1000  $\mu$ S cm<sup>-1</sup> respectively. A small quantity of sodium sulfate (1.4 to 5 mmol L<sup>-1</sup>) was added when required in order to reach a constant ionic conductivity and strength (0.02 mol L<sup>-1</sup>) for suitable comparison. The synthetic solution was representative to the average characteristics of municipal WWTP effluent.

Evolutions of Ca and Mg elements concentration were monitored using ICP-OES, SO<sub>4</sub><sup>2-</sup> evolution was followed using ionic chromatography and total inorganic carbon (TIC) was measured using TOC/TN analyzer. H<sub>2</sub>O<sub>2</sub> concentration was dosed using complexing H<sub>2</sub>O<sub>2</sub>-Ti<sup>4+</sup> spectrophotometry method. LSV, chronoamperometry and chronopotentiometry were carried out via Ametek potentiostat connected to the electrochemical cell. The electrode potential was reported against Ag-AgCl RE.

#### IV.3. Modeling

# IV.3.1. Electrolyte containing Mg<sup>2+</sup> and Ca<sup>2+</sup>

 $Mg^{2+}$  and  $Ca^{2+}$  react with  $OH^-$  to produce precipitates of  $Mg(OH)_2$  and  $Ca(OH)_2$  when the product of concentrations of both cations with  $OH^-$  exceeds the constants of solubility of  $Mg(OH)_2$  ( $K_{S,Mg(OH)_2}$ ) ( $K_{S,Mg(OH)_2} = 5.61 \times 10^{-12}$  [22, 23]) and of  $Ca(OH)_2$  ( $K_{S,Ca(OH)_2}$ ) ( $K_{S,Ca(OH)_2} = 5.00 \times 10^{-6}$  [22, 23]) respectively. Knowing  $Mg^{2+}$  and  $Ca^{2+}$  concentrations in the system, the concentration of  $OH^-$  at which its product reaches the  $K_{S,Mg(OH)_2}$  and  $K_{S,Ca(OH)_2}$  values, respectively, can be calculated. It is noted as critical  $OH^-$  concentration ( $OH^-_{crit}$ ). During the electrolysis,  $OH^-$  ions are produced on cathode by

reduction of either dissolved  $O_2$  via 4-electrons reaction (**Eq. (II.9)**) and/or water via 2-electrons reaction (**Eq. (II.10)**). Since both reactions took place in parallel on cathode in our system under the applied current densities, they have been considered as competing reactions.  $\eta_{O_2}$  and  $\eta_{H_2O}$  terms were used to proportionate the current efficiency as a result of the reduction of  $O_2$  and  $O_2$  and  $O_3$  respectively. The rate of production of  $O_3$  in mol m<sup>-2</sup> s<sup>-1</sup>) on cathode in batch mode can thus be written as in **Eq. (IV.3)**:

$$r_{\text{OH}^{-}} = + \frac{d[OH^{-}]}{dt} = \frac{v_{\text{O}_{2}} j_{\text{app}}}{n_{\text{O}_{2}} F} \eta_{\text{O}_{2}} + \frac{\alpha v_{\text{H}_{2}\text{O}} j_{\text{app}}}{n_{\text{H}_{2}\text{O}} F} \eta_{\text{H}_{2}\text{O}}$$
(IV.3)

where  $v_{0_2}$  and  $v_{H_2O}$  are the stoichiometry coefficient of OH<sup>-</sup> via reaction of reduction of O<sub>2</sub> (Eq. (II.9)) and H<sub>2</sub>O (Eq. (II.10)) respectively, while  $n_{0_2}$  and  $n_{H_2O}$  are their number of electrons involved in those faradaic reactions,  $j_{app}$  is the applied current density in A m<sup>-2</sup> and F is the Faraday constant (96485 C mol<sup>-1</sup>).  $\alpha$  is denoted as OH<sup>-</sup> hindrance factor ( $\alpha$  < 1) assigned to the faradaic production of OH<sup>-</sup> which occurs concomitantly with physical evolution of gas bubbles on cathode surface at high  $j_{app}$ . In accordance to Eq. (II.10), H<sub>2</sub> gas also evolves when H<sub>2</sub>O is reduced to OH<sup>-</sup>. H<sub>2</sub> gas evolution is more intense when higher current density value is used, according to the Faraday's law. Gas evolution on cathode surface contributes to two major consequences; firstly, less surface area could be available for mineral deposition on cathode surface [24]. Secondly, it could promote detachment of deposits occurring on the electrode [25, 26].

On the surface of cathode,  $OH^-$  ions are produced abundantly and continuously. Water is the solvent of electrolyte thus its electrochemical reduction is limitless. Meanwhile, dissolved  $O_2$  are not only consumed on cathode (**Eq.** (**II.9**)), but it is also concomitantly produced on anode through water oxidation (**Eq.** (**II.3**)).

$$2H_2O \rightarrow O_2 + 4H^+ + 4e^-$$
 (II.3)

In microfluidic configuration, the distance between cathode and anode is very small thus the replenishment of dissolved  $O_2$  could take place continuously [27]. As a result, once  $OH_{crit}^-$  is reached,  $OH_{crit}^-$  concentration is considered constant and it does not vary whilst reacting with  $Mg^{2+}$  and  $Ca^{2+}$  because the concentration of  $OH_{crit}^-$  is significantly larger and in excess compared to the cations concentrations. In addition, it is assumed that  $Mg^{2+}$  and  $Ca^{2+}$  were not reduced or oxidized by direct electrochemical reduction or oxidation, so that their oxidation state remained constant all along the electrolysis [28].

Hence, according to **Eqs. (I.11)** and **(IV.1)**, the rates of reaction of Mg<sup>2+</sup> and Ca<sup>2+</sup> (in mol m<sup>-2</sup> s<sup>-1</sup>) can be written according to **Eq. (IV.4)** and **Eq. (IV.5)** respectively:

$$r_{\text{Mg}^{2+}} = -r_{\text{Mg(OH)}_2} = -k_{\text{Mg(OH)}_2} [Mg^{2+}] [OH^-]^2$$
 (IV.4)

$$r_{\text{Ca}^{2+}} = -r_{\text{Ca(OH)}_2} = -k_{\text{Ca(OH)}_2} [Ca^{2+}][OH^-]^2$$
 (IV.5)

where  $k_{Mg(OH)_2}$  and  $k_{Ca(OH)_2}$  are the heterogeneous third order rate constants regarding the reaction of precipitation of Mg(OH)<sub>2</sub> and Ca(OH)<sub>2</sub> respectively in m<sup>7</sup> mol<sup>-2</sup> s<sup>-1</sup>, while [Mg<sup>2+</sup>], [Ca<sup>2+</sup>] and [OH<sup>-</sup>] are the molar concentration (in mol m<sup>-3</sup>) of Mg<sup>2+</sup>, Ca<sup>2+</sup> and OH<sup>-</sup> ions, respectively.

# IV.3.2. Electrolyte containing Ca<sup>2+</sup> and CO<sub>3</sub><sup>2-</sup>

 $\text{Ca}^{2+}$  reacts with  $\text{CO}_3^{2-}$  to form  $\text{CaCO}_3$  precipitates when the product of concentrations of both ions exceeds the constant of solubility of  $\text{CaCO}_3$  ( $K_{\text{S,CaCO}_3}$ ) ( $K_{\text{S,CaCO}_3} = 3.36 \times 10^{-9}$  [23, 29-31]). Supersaturation degree (SD) of  $\text{CaCO}_3$  is defined in Eq. (IV.6) and follows by its supersaturation index (SI) in Eq. (IV.7) [32, 33].

$$SD = \frac{[ca^{2+}][cO_3^{2-}]}{K_{S,CaCO_2}}$$
 (IV.6)

$$SI = log_{10}(SD) (IV.7)$$

Thermodynamically, CaCO<sub>3</sub> can precipitate when SD > 1 (or SI > 0). Carbonates concentration can be estimated from TIC measurement that corresponds to the sum of all forms of inorganic carbon (Eq. (IV.8)):

$$TIC = [H_2CO_3] + [HCO_3^-] + [CO_3^{2-}]$$
 (IV.8)

where  $H_2CO_3$  is carbonic acid and  $HCO_3^-$  is bicarbonate ion.

The inorganic carbon species are linked together by acid dissociation constants  $K_{A1}$  and  $K_{A2}$  given in Eqs. (IV.9)–(IV.10), respectively.

$$K_{\rm A1} = \frac{[H^+][HCO_3^-]}{[H_2CO_3]}$$
 (IV.9)

$$K_{A2} = \frac{[H^+][CO_3^{2-}]}{[HCO_3^-]}$$
 (IV.10)

Their ratios are therefore governed by the solution pH.  $pK_{A1}$  and  $pK_{A2}$  values are 6.3 and 10.4 [34], respectively. Under the range of our pH operating conditions and at the proximity of cathode surface, the quantity of  $H_2CO_3$  is negligible and the expression of TIC can be simplified into Eq. (IV.11).

$$TIC = [HCO_3^-] + [CO_3^{2-}]$$
 (IV.11)

 $CO_3^{2-}$  concentration is related to the concentration of  $OH^-$  produced on cathode surface by its equilibrium with  $HCO_3^-$  in agreement with **Eq.** (II.12) and **Eq.** (II.13). Hence, the evolution of interfacial  $CO_3^{2-}$  concentration ( $[CO_3^{2-}]_{int,t}$ ) can be estimated from TIC values by combining **Eqs.** (IV.10) and (IV.11) giving out an expression in **Eq.** (IV.12).

$$[CO_3^{2-}]_{\text{int,t}} = \frac{\frac{K_{A2} \times TIC_t}{K_W}}{\frac{K_W}{[OH^-]_t} + K_{A2}}$$
 (IV.12)

where  $K_{A2}$  and the concentration of interfacial  $CO_3^{2-}$ ,  $OH^-$  and TIC are expressed in mol  $L^{-1}$  while the equilibrium constant of auto-ionization of water ( $K_W$ ) is considered equaled to  $10^{-14}$  at  $25^{\circ}$ C.

According to **Eq.** (**II.13**), the rate of reaction of Ca<sup>2+</sup> and CO<sub>3</sub><sup>2-</sup> producing CaCO<sub>3</sub> can be defined as indicated in **Eq.** (**IV.13**):

$$r_{\text{Ca}^{2+}} = r_{\text{CO}_3^{2-}} = -r_{\text{CaCO}_3} = -k_{\text{CaCO}_3} [Ca^{2+}][CO_3^{2-}]_{int}$$
 (IV.13)

where  $r_i$  is in mol m<sup>-2</sup> s<sup>-1</sup>,  $k_{CaCO_3}$  is the heterogenous second order rate constant (in m<sup>4</sup> mol<sup>-1</sup> s<sup>-1</sup>) regarding the reaction of precipitation of CaCO<sub>3</sub>. Identically to the Mg<sup>2+</sup>/Ca<sup>2+</sup> system (Section IV.3.1), OH<sup>-</sup> accumulates at the vicinity of cathode during the electrolysis.

### IV.3.3. Electrolyte containing Mg<sup>2+</sup> and CO<sub>3</sub><sup>2-</sup>

In absence of Ca<sup>2+</sup>, Mg<sup>2+</sup> can precipitate in the form of Mg(OH)<sub>2</sub> upon reaching  $OH_{crit}^-$  as well as MgCO<sub>3</sub> when the product of concentrations of Mg<sup>2+</sup> and CO<sub>3</sub><sup>2-</sup> exceeds thermodynamic solubility constant of MgCO<sub>3</sub> ( $K_{S,MgCO_3}$ ) ( $K_{S,MgCO_3}$  = 1.59 × 10<sup>-8</sup> [35]). Local alkalization remains valid at the surface of cathode. Knowing the fact that Mg<sup>2+</sup> could participate in the precipitation reactions of Mg(OH)<sub>2</sub> and MgCO<sub>3</sub>, on top of kinetic law defined in Eq. (IV.4), the rate of reaction of Mg<sup>2+</sup> (in mol m<sup>-2</sup> s<sup>-1</sup>) can be rewritten in Eq. (IV.14) as follow:

$$r_{\rm Mg^{2+}} = -r_{\rm Mg(OH)_2} - r_{\rm MgCO_3} = -k_{\rm Mg(OH)_2} [Mg^{2+}] [OH^-]^2 - k_{\rm MgCO_3} [Mg^{2+}] [CO_3^{2-}]$$
 (IV.14)

where  $k_{\text{MgCO}_3}$  is the heterogeneous second order rate constant (in m<sup>4</sup> mol<sup>-1</sup> s<sup>-1</sup>) regarding the reaction of precipitation of MgCO<sub>3</sub>.

## IV.3.4. Electrolyte containing Mg<sup>2+</sup>, Ca<sup>2+</sup> and CO<sub>3</sub><sup>2-</sup>

In the matrix of electrolyte containing  $Mg^{2+}$ ,  $Ca^{2+}$  and  $CO_3^{2-}$ , both  $Mg^{2+}$  and  $Ca^{2+}$  can undergo precipitation in accordance to **Eqs. (II.11)** and **(IV.1)** and **Eqs. (II.13)** and **(IV.2)**. The phenomenon of local alkalization remains valid in this system.

In the presence of  $Mg^{2+}$ , depositing  $Mg(OH)_2$  consumes  $OH^-$  over time. The cathodic production of  $OH^-$  in **Eq. (IV.3)** is modified taking into account this event as indicated in **Eq. (IV.15)**:

$$r_{\text{OH}^{-}} = + \frac{d[OH^{-}]}{dt} = \frac{v_{\text{O}_{2}} j_{\text{app}}}{n_{\text{O}_{2}} F} \eta_{\text{O}_{2}} + \frac{\alpha v_{\text{H}_{2}\text{O}} j_{\text{app}}}{n_{\text{H}_{2}\text{O}} F} \eta_{\text{H}_{2}\text{O}} - 2 k_{\text{Mg(OH)}_{2}} [Mg^{2+}] [OH^{-}]^{2}$$
 (IV.15)

#### IV.3.5. Modeling software and fitting evaluation

Aquasim [36] was used to model the evolution of Mg<sup>2+</sup>, Ca<sup>2+</sup> and CO<sub>3</sub><sup>2-</sup>. The fitting between experimental and modeled curve was evaluated using RMSE (Eq. (III.3)), ME (Eq. (III.4)) and IOA (Eq. (III.5)) provided in Chapter 3.

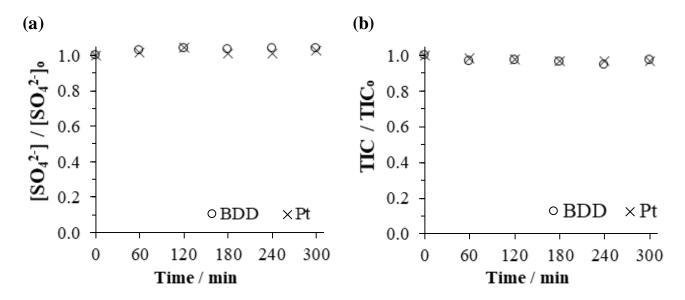
#### IV.4. Results and discussion

#### IV.4.1. Stability of anions in blank solutions using BDD or Pt anode

Preliminary experiments were conducted in the microfluidic reactor to check on the behavior of anions (SO<sub>4</sub><sup>2</sup>- and HCO<sub>3</sub>-/CO<sub>3</sub><sup>2</sup>-) during electro-oxidation in the absence of depositing elements. BDD and Pt anodes were used to verify the involvement of •OH towards any oxidation of anions. The former is well-known for its high overpotential for oxygen evolution reaction (OER). Consequently, physi-sorbed •OH are produced at anode vicinity [37, 38]. Contrastingly, the overpotential for OER with Pt anode is low. Thus •OH is chemi-sorbed on the anode surface, making the radical not available [39].

The evolution of concentration of SO<sub>4</sub><sup>2-</sup> and TIC are depicted in **Fig. IV.1**. It has been noticed that the concentration of both anions remained constant throughout the entire electro-oxidation experiments whether BDD or Pt was used as anode. First deduction that can be made is that SO<sub>4</sub><sup>2-</sup> and HCO<sub>3</sub><sup>-</sup>/CO<sub>3</sub><sup>2-</sup> anions were stable during the electro-oxidation process whatever the anode employed. It means that they will not interfere if there is occurrence of electro-precipitation on cathode surface. Secondly, under these applied conditions, the formation of oxidizing agent of •OH did not inflict towards the degradation of SO<sub>4</sub><sup>2-</sup> and HCO<sub>3</sub><sup>-</sup>/CO<sub>3</sub><sup>2-</sup> anions. Anode potential was measured to be +2.40 V/Ag-AgCl (or +2.60 V/SHE), which already hit the standard potential of formation of •OH when 4 mA cm<sup>-2</sup> is applied. The oxidation conditions were not strong enough to

oxidize SO<sub>4</sub><sup>2-</sup> into peroxydisulfates (S<sub>2</sub>O<sub>8</sub><sup>2-</sup>) [40] and CO<sub>3</sub><sup>2-</sup> into peroxydicarbonates (C<sub>2</sub>O<sub>6</sub><sup>2-</sup>) [41, 42] with SO<sub>4</sub><sup>6-</sup> and CO<sub>3</sub><sup>6-</sup> as intermediates, respectively. Moreover, anion of strong acid such as SO<sub>4</sub><sup>2-</sup> is very unlikely to form complex with alkaline-earth cations such as Ca<sup>2+</sup> and Mg<sup>2+</sup> [43]. In overall, <sup>6</sup>OH involvement in microfluidic reactor set-up did not lead to the oxidation of SO<sub>4</sub><sup>2-</sup> and HCO<sub>3</sub>-/CO<sub>3</sub><sup>2-</sup> anions in blank electrolytes during electro-oxidation experiments. In overall, the presence of <sup>6</sup>OH in our applied microfluidic reactor conditions could thus be omitted from the developed theoretical models and was confirmed further in the next sections.



**Fig. IV.1**. Evolution of concentration of (a)  $SO_4^{2-}$  (in blank solution containing sulfates only) and (b) TIC (in blank solution containing carbonates only) during the electrolysis of blank solutions using BDD (O) or Pt ( $\times$ ) as anode. Cathode: stainless steel, ionic strength of both blank solutions: 0.02 mol L<sup>-1</sup>,  $j_{app}$ : 4 mA cm<sup>-2</sup> and  $d_{elec}$ : 500  $\mu$ m.

# IV.4.2. Local alkalization on cathode surface: reactions' selectivity between reduction of dissolved O<sub>2</sub> and water

According to **Eq.** (II.9) and **Eq.** (II.10) respectively, both dissolved  $O_2$  and  $H_2O$  can undergo faradaic reduction generating  $OH^-$  (**Eq.** (II.3)). To identify the weight of contribution of dissolved  $O_2$  to produce  $OH^-$  with respect to local alkalization phenomenon, tests in presence and absence of dissolved  $O_2$  were conducted. Under the investigated micrometric distance, anodic activity might be involved in this selectivity, since  $H_2O$  could be also oxidized to  $O_2$ .

**Figure IV.2** plots the evolution of concentration of  $Mg^{2+}$ ,  $Ca^{2+}$  and TIC in the experiments with and without dissolved  $O_2$ . In both cases, the concentration of all species decreased regardless of presence of dissolved  $O_2$ . This indicates that the initial presence of dissolved  $O_2$  was insignificant towards the phenomenon of local alkalization, and consequently towards the formation of deposits on cathode surface. Notably from **Fig. IV.2(a)** and **Fig. IV.2(b)**, it has been deduced that the source of  $OH^-$  ions

came mainly from reduction of water and not from dissolved  $O_2$ . A slight decrease in the electro-precipitation of  $Ca^{2+}$  in absence of dissolved  $O_2$  was due to slightly lesser amount of  $OH^-$  participating in thermodynamic shift of  $HCO_3^-/CO_3^{2-}$  towards the formation of  $CO_3^{2-}$  (**Fig. IV.2(b)**). Thus, lesser extent of electro-precipitating  $CaCO_3$  was observed.

Under theses setups, the concentration of electrogenerated  $H_2O_2$  was measured to further study the contribution of dissolved  $O_2$  reduction producing  $OH^-$  ions in accordance with **Eq. (II.4)**.

$$O_2 + 2H_2O + 2e^- \leftrightarrow H_2O_2 + 2OH^-$$
 (II.4)

The H<sub>2</sub>O<sub>2</sub> production was found to be very small (maximum concentration went up to 1.8 μmol L<sup>-1</sup>, data not shown), whatever the applied current density. Most of H<sub>2</sub>O<sub>2</sub> concentration values fell below the limit of quantification of the method, determined to be 0.8 μmol L<sup>-1</sup>. This observation was attributed to the low H<sub>2</sub> evolution overvoltage on stainless steel cathode [44]. Similar poor production of H<sub>2</sub>O<sub>2</sub> on stainless cathode was reported by Sopaj et al. [45]. They measured approximately 40 μmol L<sup>-1</sup> of H<sub>2</sub>O<sub>2</sub> under their experimental setup when 4.8 mA cm<sup>-2</sup> was applied [45]. Unlike the configuration presented in our work, compressed air was continuously bubbled under their experimental setup to supply dissolved O<sub>2</sub> to promote the Fenton reaction, which explains the difference in measured H<sub>2</sub>O<sub>2</sub> concentration.

 $\eta_{\rm O_2}$  and  $\eta_{\rm H_2O}$  in **Eq. (IV.3)** and **Eq. (IV.15)** were estimated via **Eqs. (IV.16)-(IV.17)** at each value of current density investigated in this work.

$$\eta_{O_2} = \frac{n_e F N_{O_2}}{I t} \tag{IV.16}$$

$$\eta_{\rm H_2O} = 1 - \eta_{\rm O_2}$$
 (IV.17)

where  $n_e$  is the number of electrons exchanged in **Eq.** (II.4) (i.e., 2),  $N_{O_2}$  is the quantity of dissolved  $O_2$  (in mol) reduced to  $H_2O_2$  in **Eq.** (II.4), I is the applied current (in A) and t is the electrolysis time (in s).

At 4 mA cm<sup>-2</sup> (or 200 mA), the highest concentration of  $H_2O_2$  (1.8  $\mu$ mol L<sup>-1</sup>) was found after 1 h of electrolysis. In 500 mL of electrolyte, knowing the fact that same number of moles of dissolved  $O_2$  ( $N_{O_2}$ , in mol) is consumed to produce the number of moles of  $H_2O_2$  ( $N_{H_2O_2}$ , in mol):

$$\eta_{O_2} = \frac{n_e F N_{O_2}}{I t} = \frac{2 \times 96485 \times 0.9 \times 10^{-6}}{0.2 \times 3600} \times 100 = 0.02\%$$

$$\eta_{H_2O} = 100 - \eta_{O_2} = 99.98\%$$

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While at 0.4 mA cm<sup>-2</sup> (or 20 mA), the highest concentration of  $H_2O_2$  (1.18  $\mu$ mol  $L^{-1}$ ) was found after 1 h of electrolysis.

$$\eta_{\text{O}_2} = \frac{2 \times 96485 \times 0.59 \times 10^{-6}}{0.02 \times 3600} \times 100 = 0.16\%$$

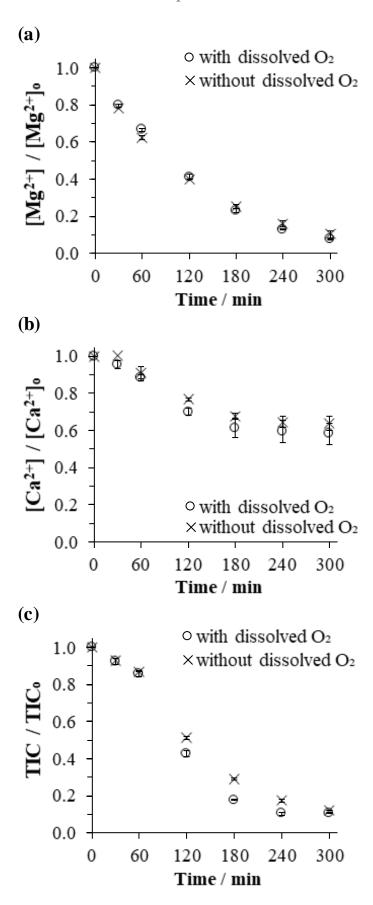
$$\eta_{\text{H}_2\text{O}} = 100 - \eta_{\text{O}_2} = 99.84\%$$

Their values are regrouped in **Table IV.1**.

**Table IV.1**. Estimation of  $\eta_{0_2}$  and  $\eta_{H_20}$  in faradaic production of OH<sup>-</sup> on cathode.

j <sub>app</sub> / mA cm <sup>-2</sup>	$N_{O_2}$ / $\mu$ mol	$CE_{0_2}$ / %	CE <sub>H2O</sub> / %
0.4	0.6	0.16	99.84
4	0.9	0.02	99.98

To conclude, local alkalization on cathode surface principally occurred due to the reduction of  $H_2O$  with only slight involvement of dissolved  $O_2$  under our experimental setup. The latter was electrocatalytically reduced to  $OH^-$  with negligible production of  $H_2O_2$  on stainless steel. Our proposed model takes into consideration this observed selectivity given by higher faradaic yield of  $\eta_{H_2O}$  in producing  $OH^-$  as deduced in **Table IV.1** when applied to **Eq. (IV.3)** and **Eq. (IV.15)**.



**Fig. IV.2**. Evolution of concentration of (a)  $Mg^{2+}$ , (b)  $Ca^{2+}$  and (c) TIC during the electrolysis of artificial effluent containing 150 mg L<sup>-1</sup> of  $Ca^{2+}$ , 5 mg L<sup>-1</sup> of  $Mg^{2+}$  and 60 mg-C L<sup>-1</sup> of  $HCO_3^{-}/CO_3^{2-}$  in absence (×) (via bubbling with  $N_2$  gas) and presence (O) of dissolved  $O_2$  throughout the experiment. Anode: BDD, cathode: stainless steel,  $j_{app}$ : 4 mA cm<sup>-2</sup> and  $d_{elec}$ : 500 μm.

#### IV.4.3. Influence of electromigration of ionic species

Although calcium and magnesium deposits are formed on the surface of cathode during electrolysis,  $Ca^{2+}$  and  $Mg^{2+}$  do not participate in direct electron transfer to yield Ca metal (Ca(s)) and Mg metal (Mg(s)) respectively. The reduction of  $Ca^{2+}$  and  $Mg^{2+}$  each by two electrons need to overcome a strong thermodynamic barrier ( $E_{Ca^{2+}/Ca_{(s)}}^0$  = -2.84 V/SHE and  $E_{Mg^{2+}/Mg_{(s)}}^0$  = -2.356 V/SHE [46]), and these strong reduction conditions are not involved in the present study (cathode potential determined to be -1.34 V/SHE). Instead, they reacted chemically with respective counter-ions on the surface of cathode, producing scaling when solubility constant of precipitates was achieved. Despite not being electroactive elements in regard to driving force of potential difference across electrochemical reactor,  $Ca^{2+}$  and  $Mg^{2+}$  mobility towards cathode surface might be involved in the displacement of ionic species jointly present in the bulk. In agreement with the Nernst-Planck equation,  $Ca^{2+}$  and  $Ca^{2+}$  and  $Ca^{2+}$  were transported by forced convection thanks to electrolyte flow in-between electrodes. Under stationary conditions, they underwent diffusion from electrolyte bulk through finite boundary layer (i.e. diffusion layer), towards reactive zone at the proximity of cathode surface as the consequence of local alkalization [47-49].

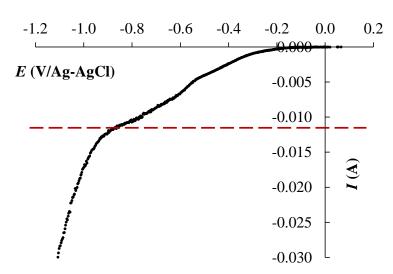
The initial limiting current density  $(j_{\text{lim},0})$  for the adopted electrochemical design operating with the working synthetic effluent is calculated using **Eq. (IV.18)** [50, 51].

$$I_{\lim} = n_{\rm e} F S k_{\rm m} C_{\rm SOL} \tag{IV.18}$$

where  $n_e$  is the number of electrons participating in the limiting reaction of **Eq. (II.9)**, S the effective surface of cathode (m<sup>2</sup>),  $k_m$  the mass transfer coefficient for the assigned electrochemical cell from the bulk towards the cathode (m s<sup>-1</sup>) and  $C_{SOL}$  the molar concentration of considered reactive species in the bulk (mol m<sup>-3</sup>).

Reaction of dissolved O<sub>2</sub> reduction via 4-electrons producing OH<sup>-</sup> (Eq. (II.9)) was selected as limiting step in agreement to the very high  $\eta_{\rm H_2O}$  value.  $n_{\rm e}$  is thus equal to 4. The stainless steel cathode possesses an effective surface area of 50 cm<sup>-2</sup>.  $k_{\rm m}$  equals to 2.33 × 10<sup>-5</sup> m s<sup>-1</sup> (Chapter 5). The concentration of dissolved O<sub>2</sub> was taken at 25°C, which is 2.81 × 10<sup>-4</sup> mol L<sup>-1</sup>. Computing Eq. (IV.18) gave  $I_{\rm lim,0}$  equals to 0.013 A, which lead to a  $j_{\rm lim,0}$  of 0.25 mA cm<sup>-2</sup>.  $I_{\rm lim,0}$  value was ascertained experimentally with the plateau of limiting current density spotted on the voltammogram of the experimental setup. Therefore, the two current densities investigated in this study were both above the evaluated  $j_{\rm lim,0}$ . As a result, the electrochemical reactions occurring on the surface of cathode were considered limited by mass transfer.

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**Fig. IV.3**. Linear scan voltammetry (LSV) on stainless steel working electrode under cathodic polarization at a scan rate of 10 mV s<sup>-1</sup>. BDD was used as CE.  $d_{\rm elec}$ : 500  $\mu$ m. 0.5 L of electrolyte containing 150 mg L<sup>-1</sup> Ca<sup>2+</sup>, 5 mg L<sup>-1</sup> Mg<sup>2+</sup> and carbonates of 60 mg-C L<sup>-1</sup> with ionic strength of 0.02 mol L<sup>-1</sup>.

In this section, under constant hydrodynamic parameters, the influence of electromigration of ionic species was investigated by varying the ionic strength of artificial effluent using Na<sub>2</sub>SO<sub>4</sub> as supporting electrolyte. **Table IV.2** summarizes the values of ionic transference number for the different cases. They were calculated using **Eqs. (IV.19)-(IV.20)** [52] under the assumption that our electrolytic system complied with the restriction of infinite dilution where interaction between ions was not occurring.

$$t_i = \frac{z_i c_i u_i}{\sum_i z_i c_i u_i} \tag{IV.19}$$

with

$$u_i = \frac{\lambda_i}{z_i F} \tag{IV.20}$$

where  $t_i$  is the transference number,  $z_i$  the ionic charges,  $u_i$  the ionic mobility (in m<sup>2</sup> s<sup>-1</sup> V<sup>-1</sup>),  $C_i$  the molar concentrations of ionic species (in mol L<sup>-1</sup>) and  $\lambda_i$  the ionic molar conductivity (in S m<sup>2</sup> mol<sup>-1</sup>).

**Table IV.2**. Ionic transference number for artificial effluent containing 150 mg L<sup>-1</sup> of Ca<sup>2+</sup>, 5 mg L<sup>-1</sup> of Mg<sup>2+</sup> and 60 mg-C L<sup>-1</sup> of TIC without and with addition of Na<sub>2</sub>SO<sub>4</sub> at different concentrations.

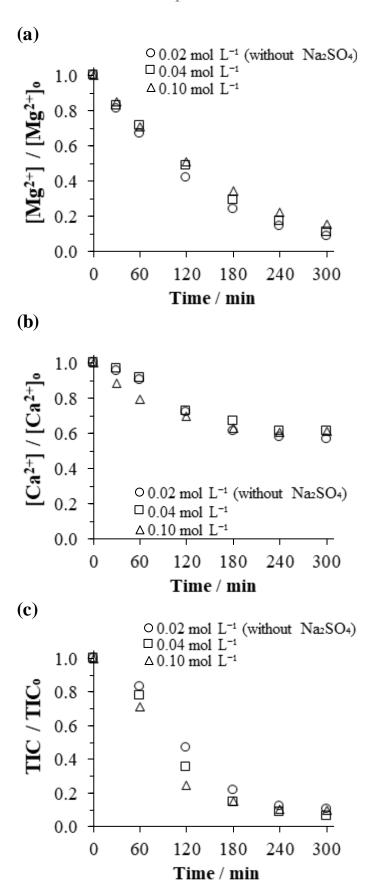
Ionic strength /	Ionic transference number				
mol L <sup>-1</sup>	Ca <sup>2+</sup>	Mg <sup>2+</sup>	Na <sup>+</sup>	HCO <sub>3</sub>	SO <sub>4</sub> <sup>2-</sup>
0.02 (without Na <sub>2</sub> SO <sub>4</sub> )	0.218	0.011	0.123	0.339	0.309
0.04	0.120	0.006	0.241	0.187	0.447
0.10	0.050	0.002	0.325	0.077	0.545

As can be seen from **Table IV.2**, the more  $Na_2SO_4$  was added to increase the ionic strength of artificial effluent, the more ionic mobility of  $Ca^{2+}$ ,  $Mg^{2+}$  and  $HCO_3^{-}/CO_3^{2-}$  were minimized by increasing transport activity of  $Na^+$  and  $SO_4^{2-}$ .

Fig. IV.4 depicts the trend of evolution of concentrations of Mg<sup>2+</sup>, Ca<sup>2+</sup> and TIC when the investigated artificial effluent had different ionic strengths. The concentration of each ion decreased over time regardless of value of ionic strength. It seemed to have very little impact on the variation of the concentrations of Ca<sup>2+</sup> and TIC as previously suggested by Tlili et al., where migration might play some role with respect to only more concentrated species in the system [47]. Standard deviations were calculated between results obtained with the electrolytes with enhanced ionic mobility in comparison to the one without supporting electrolyte. The average values are listed in Table IV.3. The two highest deviations were obtained in Mg<sup>2+</sup> and TIC concentrations with evaluated deviation as low as 0.049 and 0.042, respectively. Therefore, the ionic strength of electrolyte had minor effect within the range of studied electrolyte concentrations towards the occurring mineral deposition on cathode surface. Electromigration of ionic species other than Ca<sup>2+</sup>, Mg<sup>2+</sup> and HCO<sub>3</sub><sup>-</sup>/CO<sub>3</sub><sup>2-</sup> did not influence the mobility of the precipitating elements in-between electrodes inside microfluidic electrochemical reactor. Mass transfer by diffusion was then the rate-limiting factor. Based on these results, the effect of electromigration could be omitted from the theoretical model proposed in Section IV.3. The addition of small quantity of supporting electrolyte (from 1.4 to 5 mmol L<sup>-1</sup>) to achieve constant ionic strength across different investigated matrices of electrolyte will not influence the kinetics of mineral electro-precipitation in experimental results presented onwards.

**Table IV.3**. Average standard deviations calculated for electrolytes having enhanced ionic strengths (0.04 and 0.10 mol L<sup>-1</sup>).

Floment	Average standard deviation			
Element —	0.04 mol L <sup>-1</sup>	0.10 mol L <sup>-1</sup>		
$ m Mg^{2+}$	0.030	0.049		
Ca <sup>2+</sup>	0.018	0.021		
TIC	0.029	0.042		



**Fig. IV.4**. Evolution of concentrations of (a)  $Mg^{2+}$ , (b)  $Ca^{2+}$  and (c) TIC during the electrolysis of artificial effluents bearing different ionic strengths of 0.02 (O) (without the addition of supporting electrolyte), 0.04 ( $\square$ ) and 0.10 ( $\triangle$ ) mol L<sup>-1</sup>. Anode: BDD, cathode: stainless steel,  $j_{app}$ : 4 mA cm<sup>-2</sup> and  $d_{elec}$ : 500 μm.

#### IV.4.4. Influence of matrix of electrolyte

#### IV.4.4.1. Influence of CO<sub>3</sub><sup>2-</sup> towards the mineral electro-precipitation

Electrolysis of electrolyte containing Mg<sup>2+</sup> and Ca<sup>2+</sup>, in presence and absence of CO<sub>3</sub><sup>2-</sup>, in the microfluidic flow-by reactor were carried out by applying 0.4 and 4 mA cm<sup>-2</sup> and the results are plotted in Fig. IV.5 and Fig. IV.6. From Fig. IV.5, it could firstly be observed that Ca<sup>2+</sup> did not precipitate in the electrolyte without carbonate at both current densities. It precipitated only in presence of carbonate. Thus, it has been concluded that Ca(OH)<sub>2</sub> was not produced throughout these series of experiment and Ca<sup>2+</sup> only electro-precipitated to form CaCO<sub>3</sub>. This observation can be supported by the value of thermodynamic constant of solubility of Ca(OH)<sub>2</sub> that is high  $(5.00 \times 10^{-6}$ [22, 23]) - thus highly soluble -, hence the interfacial pH never reached the OH<sub>crit</sub> (i.e. 36.6 mol m<sup>-3</sup>, pH = 12.6) to precipitate with  $Ca^{2+}$ . Secondly, according to Fig. IV.6,  $Mg^{2+}$  only precipitated at 4 mA cm<sup>-2</sup> and no Mg<sup>2+</sup>-based deposition occurred at 0.4 mA cm<sup>-2</sup>. At low current density, since no decrease of  $Mg^{2+}$  was observed, it can be concluded that  $Mg(OH)_2$  was not electro-precipitated. Same remark was given by Deslouis et al. [14] where they did not observe precipitation of Mg(OH)2 when interfacial pH of 9.3 was not yet reached. Our result tells that there was a plateau at 4 mA cm<sup>-2</sup> at the very beginning of electrolysis up to 7 min before Mg<sup>2+</sup> started to deposit forming Mg(OH)<sub>2</sub>, as soon as  $OH_{crit}^-$  (0.16 mol m<sup>-3</sup>, interfacial pH = 10.2) was reached (Fig. IV.6). Then, during the first hour of electrolysis at 4 mA cm<sup>-2</sup>, a slightly higher kinetics of Mg(OH)<sub>2</sub> deposition was observed in the electrolyte without carbonate. In the presence of CO<sub>3</sub><sup>2</sup>-, there was perhaps a weak competition between Mg(OH)<sub>2</sub> and CaCO<sub>3</sub> for active site on the surface of cathode [20, 53]. However, it was not accounted for MgCO<sub>3</sub>, otherwise the kinetics of precipitation with the electrolyte containing CO<sub>3</sub><sup>2</sup>would be faster and not slower than the one without CO<sub>3</sub><sup>2</sup>. To conclude, Mg<sup>2+</sup> electro-precipitated independently with regard to carbonate ions.

Moreover, it can be seen in **Fig. IV.5** that CaCO<sub>3</sub> was already deposited at low current density, unlike Mg(OH)<sub>2</sub> even though the thermodynamic constant of solubility of Mg(OH)<sub>2</sub> is lower than that of CaCO<sub>3</sub>. Similar observation was reported by Okstad et al. [54]. The higher initial concentration of Ca<sup>2+</sup> - implemented to simulate the composition of reclaimed wastewater from urban WWTPs - can be one of the reasons of this observation, whereby supersaturation degree was already high. On top of that, experimental results showed that 7.2% more CaCO<sub>3</sub> deposition occurred at 0.4 mA cm<sup>-2</sup> in comparison to higher applied current density of 4 mA cm<sup>-2</sup>. To our understanding, once CaCO<sub>3</sub> electro-precipitation started to occur, it took place regardless of applied current density as long as the current density was equal or higher than the starting point of the nucleation of CaCO<sub>3</sub> crystal. In case of low and high applied current densities investigated in this work, CaCO<sub>3</sub> already formed at the low

 $j_{\rm app}$  (0.4 mA cm<sup>-2</sup>). It would behave similarly at higher applied current density except that within the latter, oxidation and reduction of solvent on the surface of electrodes were occurring at higher rates leading to higher amount of electro-generated gas. Given the fact that at 0.4 mA cm<sup>-2</sup> it was already exceeding the limiting current density (0.25 mA cm<sup>-2</sup>, see Section IV.4.3), gas production was therefore occurring at both applied conditions. It means that the influence of gas evolution was taking place concomitantly with the electro-precipitation on cathode surface [55], but in an antagonist way according to the level of gas production. At 0.4 mA cm<sup>-2</sup>, the lower evolution of gas on the surface of electrode could enhance the mass transfer of species towards the electrodes [56-59]. This enhancement would be attributed to firstly micro-convection, even at small value of current density [56], due to evolving and departing gas bubbles from the electrode surface. Secondly, forced convection or macro-convection owing to movement and displacement of bubbles over the electrode surface could also occur [57-59]. Therefore, transport of Ca<sup>2+</sup> and Mg<sup>2+</sup> towards cathode surface for precipitating reaction were enhanced at the low  $j_{app}$  condition. This could explain why slightly more  $CaCO_3$  was obtained at lower  $j_{app}$ . Contrastingly, higher intensity of gas evolution is expected over the surface of cathode at high  $j_{app}$  (4 mA cm<sup>-2</sup>). Therefore, bubble coverage at the electrode surface should have increased as described by Vogt et al. [24]. Therefore, the more intense gas evolution at higher  $j_{app}$  could have disturbed the layer of CaCO<sub>3</sub> nuclei and in-parallel promoted the detachment of CaCO<sub>3</sub> deposit [25, 26, 44]. Consequently, less electro-precipitation of CaCO<sub>3</sub> was observed at higher  $j_{app}$ .

Furthermore, in presence of Ca<sup>2+</sup> (150 mg L<sup>-1</sup>) and carbonates (TIC of 60 mg-C L<sup>-1</sup>), SI defined in Eq. (IV.7) equaled to 0.89 in the initial electrolyte. It means that Ca<sup>2+</sup> and CO<sub>3</sub><sup>2-</sup> concentrations in the bulk were already above their equilibrium state in water (where SI = 0) initially [32]. From the thermodynamic point of view, CaCO<sub>3</sub> precipitation in the bulk and scaling on metal substrate could occur. However, the induction time for homogenous and heterogeneous CaCO<sub>3</sub> nucleation varies in function of physicochemical component of bulk liquid as well as metal substrate [13, 33, 60]. According to Gabrielli et al., thermodynamics requires a 40-fold supersaturation before the precipitation in the bulk can be observed, i.e., equivalent to SI value around 1.6 [2, 60, 61]. It has also been demonstrated that under low supersaturation conditions, very long induction time (12 h) was required before observing scaling on metal substrate [62]. This might be one of the reasons sluggish kinetics of electro-precipitation of CaCO<sub>3</sub> was observed at the very beginning of electrolysis in Fig. IV.5. With respect to the evolution of experimental TIC concentrations (Fig. IV.7(a-d)) as well as to the modeled interfacial CO<sub>3</sub><sup>2-</sup> concentrations illustrated in Fig. IV.8, this slow initial kinetics were also noticed. From Fig. IV.8, it can be noticed that on cathode surface, CO<sub>3</sub><sup>2-</sup> departed from its initial bulk concentration and rapidly increased owing to electrogenerated OH- once the electrolysis started. With increasing quantity of interfacial CO<sub>3</sub><sup>2-</sup>, more CO<sub>3</sub><sup>2-</sup> was readily available to react with Ca<sup>2+</sup> to

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form CaCO3 scaling. The supersaturation degree quickly increased, thus promoting the deposition of CaCO<sub>3</sub>. The concentration of interfacial CO<sub>3</sub><sup>2</sup>- kept increasing as long as the rate of production of OH<sup>-</sup> was superior to the rate of production of CaCO<sub>3</sub> (or CO<sub>3</sub><sup>2-</sup> consumption at the interface). During this period, more CO<sub>3</sub><sup>2</sup>- was shifted from HCO<sub>3</sub><sup>-</sup> form since the OH<sup>-</sup> concentration also increased, the phenomenon that contributed towards local alkalization. A peak was attained when the rate of OHproduction equaled that of CO<sub>3</sub><sup>2</sup>- consumed. Beyond this peak, the rate of electro-precipitation of CaCO<sub>3</sub> declined due to decreasing concentration of both Ca<sup>2+</sup> and CO<sub>3</sub><sup>2-</sup> in the electrolyte at later stage of electrolysis as can be observed in Fig. IV.5. When comparing the evolution of interfacial CO<sub>3</sub><sup>2-</sup> concentration modeled at 4 mA cm<sup>-2</sup> (Fig. IV.8(a)) against 0.4 mA cm<sup>-2</sup> (Fig. IV.8(b)), three remarks can be made; firstly, longer time was needed to reach the peak at  $r_{\rm OH^-} = r_{\rm CaCO_3}$ . Secondly, smaller concentration of interfacial CO<sub>3</sub><sup>2</sup>- was produced at higher current density and thirdly, CaCO<sub>3</sub> electro-precipitation occurred at relatively slower rate after the peak of equivalence. The involvement of gas evolution might explain the described behaviors. As above-mentioned, intensified evolving H<sub>2</sub> gas on cathode surface at higher  $j_{app}$  decreased significantly the effective surface area of cathode even though the OH<sup>-</sup> production was enhanced in the meantime. Consequently, longer buffer time was observed in Fig. IV.5(a) (inset curve) at the beginning of electrolysis in comparison to Fig. IV.5(b). Lesser amount of interfacial CO<sub>3</sub><sup>2</sup> was also expected at the proximity of cathode surface due to the intervention of evolving gas. This disturbance on cathode took place throughout the electrolysis and consequently, plausible detachment of CaCO<sub>3</sub> scaling occurred along the way. In overall, slower kinetics of CaCO<sub>3</sub> deposition was observed under high current density, affirmed by the trend depicted in Fig. IV.5.

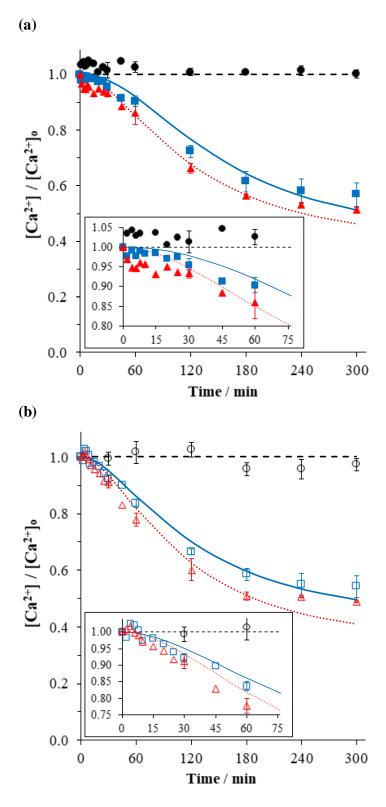
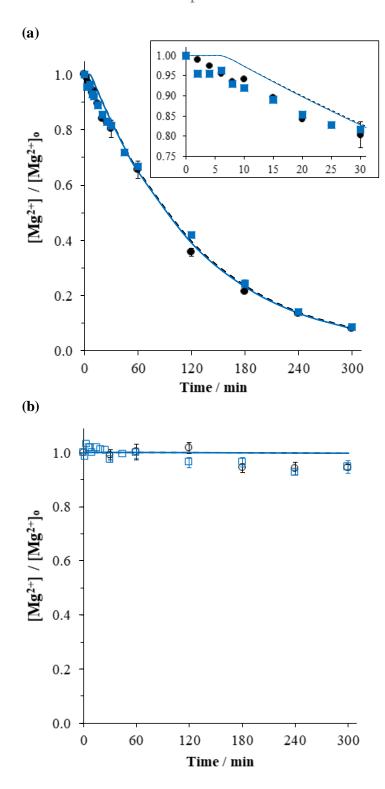
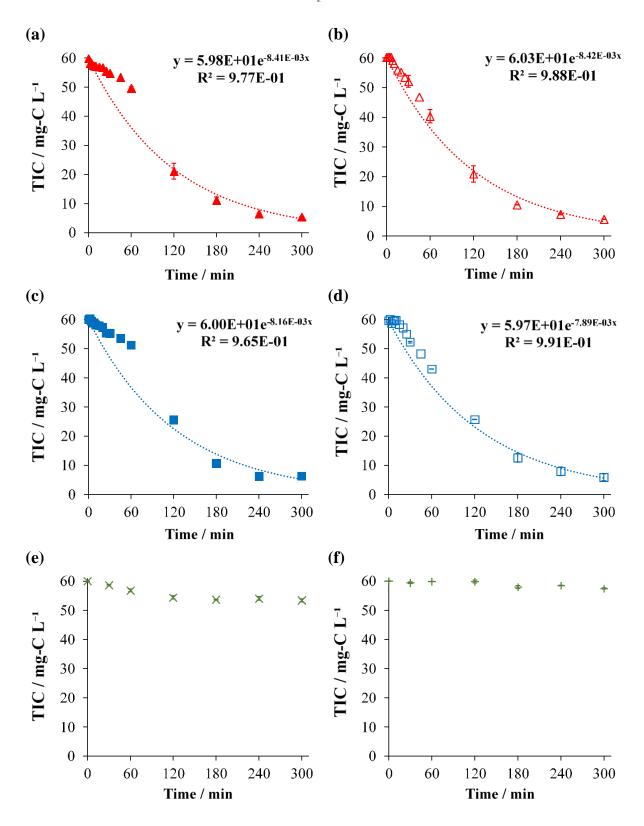


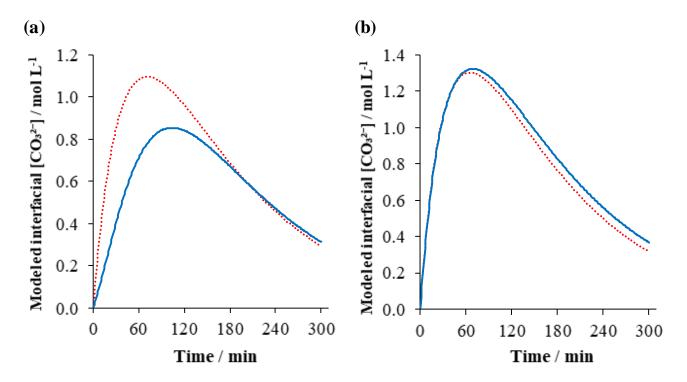
Fig. IV.5. Evolution of Ca<sup>2+</sup> concentration during the electrolysis of different matrices of electrolyte ((♠, O): Ca<sup>2+</sup> and Mg<sup>2+</sup> without CO<sub>3</sub><sup>2-</sup>, (♠, □): Ca<sup>2+</sup>, Mg<sup>2+</sup> and CO<sub>3</sub><sup>2-</sup> and (♠, △): Ca<sup>2+</sup> and CO<sub>3</sub><sup>2-</sup> without Mg<sup>2+</sup>) at (a) 4 mA cm<sup>-2</sup> (♠, ♠, △) and (b) 0.4 mA cm<sup>-2</sup> (♠, □, △). (- - -): model curve for Ca<sup>2+</sup> and Mg<sup>2+</sup> electrolyte without CO<sub>3</sub><sup>2-</sup>, (——): model curve for Ca<sup>2+</sup>, Mg<sup>2+</sup> and CO<sub>3</sub><sup>2-</sup> electrolyte and (——): model curve for Ca<sup>2+</sup> and CO<sub>3</sub><sup>2-</sup> without Mg<sup>2+</sup> electrolyte. Inset: zoom of the initial kinetics evolution of Ca<sup>2+</sup>. Anode: BDD, cathode: stainless steel and d<sub>elec</sub>: 500 μm.



**Fig. IV.6**. Evolution of Mg<sup>2+</sup> concentration during the electrolysis of different matrices of electrolyte (( $\bigcirc$ , O): Ca<sup>2+</sup> and Mg<sup>2+</sup> without CO<sub>3</sub><sup>2-</sup> and ( $\bigcirc$ ,  $\square$ ): Ca<sup>2+</sup>, Mg<sup>2+</sup> and CO<sub>3</sub><sup>2-</sup>) at (**a**) 4 mA cm<sup>-2</sup> ( $\bigcirc$ ,  $\square$ ) and (**b**) 0.4 mA cm<sup>-2</sup> ( $\bigcirc$ ,  $\square$ ). ( $\bigcirc$ ). ( $\bigcirc$ ). ( $\bigcirc$ ) model curve for Ca<sup>2+</sup> and Mg<sup>2+</sup> electrolyte without CO<sub>3</sub><sup>2-</sup> and ( $\bigcirc$ ): model curve for Ca<sup>2+</sup>, Mg<sup>2+</sup> and CO<sub>3</sub><sup>2-</sup> electrolyte. **Inset**: zoom of the initial kinetics evolution of Mg<sup>2+</sup>. Anode: BDD, cathode: stainless steel and  $d_{\text{elec}}$ : 500 μm.



**Fig. IV.7**. Evolution of experimental TIC concentration over time over the course of electrolysis at 4 mA cm<sup>-2</sup> (**a**, **c**, **e**) and 0.4 mA cm<sup>-2</sup> (**b**, **d**, **f**) using ( $\blacksquare$ ,  $\square$ ): Ca<sup>2+</sup>, Mg<sup>2+</sup> and CO<sub>3</sub><sup>2-</sup> electrolyte, ( $\triangle$ ,  $\triangle$ ): Ca<sup>2+</sup> and CO<sub>3</sub><sup>2-</sup> without Mg<sup>2+</sup> electrolyte and ( $\times$ , +) Mg<sup>2+</sup> and CO<sub>3</sub><sup>2-</sup> without Ca<sup>2+</sup> electrolyte. Dotted line corresponds to pseudo-first order kinetics of TIC decay. Anode: BDD, cathode: stainless steel,  $d_{\text{elec}}$ : 500  $\mu$ m.



**Fig. IV.8**. Modeled concentration of interfacial  $CO_3^{2-}$  concentration during the electrolysis of electrolyte containing  $Ca^{2+}$ ,  $Mg^{2+}$  and  $CO_3^{2-}$  (——) and electrolyte containing:  $Ca^{2+}$  and  $CO_3^{2-}$  without  $Mg^{2+}$  (——) under the polarization of (a) 4 mA cm<sup>-2</sup> and (b) 0.4 mA cm<sup>-2</sup>. Anode: BDD, cathode: stainless steel and  $d_{elec}$ : 500 μm.

## IV.4.4.2. Influence of Mg<sup>2+</sup> towards the mineral electro-precipitation

To investigate the role of Mg<sup>2+</sup> towards CaCO<sub>3</sub> deposition in microfluidic reactor, two electrolytes, both consisting of Ca<sup>2+</sup> and CO<sub>3</sub><sup>2-</sup> but only one contained Mg<sup>2+</sup>, were compared. According to Fig. IV.5, at both applied current densities, concentrations of Ca<sup>2+</sup> decreased over time. In both cases, higher degree of CaCO<sub>3</sub> electro-precipitation (12%) was observed in the system in absence of Mg<sup>2+</sup>. This difference was even more significant when the electrochemical cell was polarized with lower current density where the interface was less perturbed by evolving gas (Fig. IV.5(b)). This finding can be attributed to the inhibition effect of Mg<sup>2+</sup> during CaCO<sub>3</sub> deposition [9, 14, 18, 25, 32]. Mg<sup>2+</sup> can either incorporate into the CaCO<sub>3</sub> lattices lowering their growth kinetics on cathode surface [9, 32, 63], or, a thin layer of magnesian gel-like film [25, 53, 64] - which was formed prior to the CaCO<sub>3</sub> deposition [21, 64] - partially limited the surface reactivity of cathode towards the scaling of CaCO<sub>3</sub>. Nonetheless, it has been reported that CaCO<sub>3</sub> crystal can still develop and grow on the layer of Mg(OH)<sub>2</sub> as confirmed by mixture of brucite and aragonite crystals from microscopic techniques [25, 64-66]. With regard to the competing reaction of gas evolution on the surface of cathode at higher current density, it remained true while working with electrolyte without Mg<sup>2+</sup>. More intense H<sub>2</sub> gas evolution led to less CaCO<sub>3</sub> electro-precipitation (Fig. IV.5) as previously described. A slight difference was found in terms of CaCO<sub>3</sub> crystallography according to literature. When cathode potential was relatively low in the region of solvent reduction, CaCO<sub>3</sub> deposit has been reported to crystallize under cubic calcite allotropy. Whilst in presence of Mg<sup>2+</sup>, CaCO<sub>3</sub> takes aragonite form [2, 7, 21, 25, 44]. Under strong evolving gas activity, porous structure of CaCO<sub>3</sub> deposit has been reported [19-21, 26, 67].

In addition, the trends of modeled interfacial  $CO_3^{2-}$  concentration depicted on Fig. IV.8 complement the deduction made on the inhibiting role of  $Mg^{2+}$  towards  $CaCO_3$  electro-precipitation. At higher  $j_{app}$ (4 mA cm<sup>-2</sup>, Fig. IV.8(a)), Mg(OH)<sub>2</sub> deposit was formed in the electrolyte containing Mg<sup>2+</sup>. The formation of Mg(OH)<sub>2</sub> deposit partly consumed interfacial OH. As a result,  $r_{OH}$  decreased as accounted in the model (Eq. (IV.21)). Together with H<sub>2</sub> gas evolution at higher current polarization, the increment of interfacial CO<sub>3</sub><sup>2-</sup> concentration was slower in the electrolyte in presence of Mg<sup>2+</sup> relatively to the one without Mg<sup>2+</sup>. Smaller interfacial CO<sub>3</sub><sup>2-</sup> concentration was expected at cathode surface therefore lowering kinetic rate of CaCO<sub>3</sub> electro-precipitation. Contrastingly, Mg(OH)<sub>2</sub> electro-precipitation did not take place at lower  $j_{app}$ . Gas evolution was significantly lower at 0.4 mA cm<sup>-2</sup>. Consequently, the evolution of modeled interfacial CO<sub>3</sub><sup>2-</sup> concentrations (**Fig. IV.8(b**)) for both electrolytes in absence and presence of Mg<sup>2+</sup> were quite similar. It means that the availability of interfacial CO32- on cathode surface was about the same for either electrolyte. Hence, it was then confirmed from Fig. IV.5(b) that the lower growth rate of CaCO<sub>3</sub> deposition obtained in the presence of Mg<sup>2+</sup> at lower current density was due to the decrease of supersaturation degree of CaCO<sub>3</sub> in the presence of Mg<sup>2+</sup> [68]. As previously discussed, it can adsorb or incorporate into CaCO<sub>3</sub> crystal lattice inhibiting its growth during the electrolysis [32, 63].

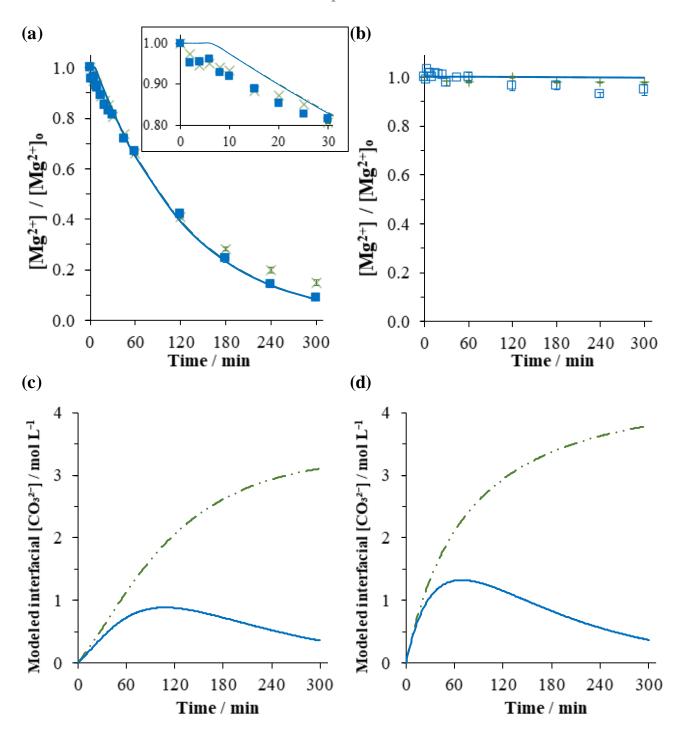
# IV.4.4.3. Influence of Ca<sup>2+</sup> towards the mineral electro-precipitation

**Fig. IV.9** illustrates the behavior of mineral deposition during electrolysis in the presence or absence of  $Ca^{2+}$  at 4 and 0.4 mA cm<sup>-2</sup>. From **Fig. IV.9(a)**, it was observed that  $Mg^{2+}$  underwent electroprecipitation independently of  $Ca^{2+}$ . The evolution of concentration of  $Mg^{2+}$  in presence of  $Ca^{2+}$  followed exactly the trend given in absence of  $Ca^{2+}$ . This observation indicated indifferent role of  $Ca^{2+}$  towards the deposition of  $Mg^{2+}$ . From **Fig. IV.9(b)**, it was noticed that  $Mg^{2+}$  did not precipitate at lower applied current density (0.4 mA cm<sup>-2</sup>) in both matrices of electrolyte. The evolution of TIC in **Fig. IV.7(e-f)** suggested that the concentration of  $CO_3^{2-}$  declined negligibly over time in the matrix without  $Ca^{2+}$  relatively to the behavior in the presence of  $Ca^{2+}$ . This tendency coupled with the observation made in **Fig. IV.9(a)** allowed deducing on the absence of electro-precipitation of  $MgCO_3$  in our operated microfluidic cell configuration. In addition, considering the fact that  $CO_3^{2-}$  did not precipitate in the absence of  $Ca^{2+}$ , the results in this section corroborated with those in **Section IV.4.4.1** in which  $CO_3^{2-}$  only underwent precipitation with  $Ca^{2+}$  to form  $CaCO_3$  deposit. The evolution of modeled interfacial  $CO_3^{2-}$  concentration in the case without  $Ca^{2+}$  plotted in **Figs. IV.9(c)** 

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and **IV.9(d)** might as well come into similar conclusion. Interfacial  $CO_3^{2-}$  concentration increased in conjunction to  $OH^-$  production on electrode surface. No  $CO_3^{2-}$  was consumed at the vicinity of cathode in the absence of  $Ca^{2+}$  allowing a continuous and quicker shift into  $CO_3^{2-}$  form at cathode interface. The interfacial  $CO_3^{2-}$  concentration arrived at a quasi-plateau at the end of electrolysis when the majority of carbon was converted into  $CO_3^{2-}$  form. Slower rate of  $OH^-$  production at higher applied current was due to involvement of  $H_2$  evolution activity.

To summarize, Mg<sup>2+</sup> deposited under the form of Mg(OH)<sub>2</sub> and it was not influenced by the presence of Ca<sup>2+</sup> nor CO<sub>3</sub><sup>2-</sup>. Only applied current density influenced the electro-precipitation of Mg(OH)<sub>2</sub> as already described in **Section IV.4.4.1**. According to Bénézeth et al. [35], MgCO<sub>3</sub> or commonly known as magnesite, possesses very slow kinetics of crystallization. Other hydrated Mg-carbonates forms are plausible owing to slow precipitation kinetics of the former, but they are rather unstable due to strong hydrated character of Mg<sup>2+</sup> [35, 69, 70]. As a result, direct electro-precipitation of MgCO<sub>3</sub> would be realizable but only under stringent conditions, i.e., elevated departing supersaturated solution, elevated temperature and very high CO<sub>2</sub> partial pressure [71]. Hence, based on our experimental results supported by the work of other authors, MgCO<sub>3</sub> was not formed under our experimental conditions.

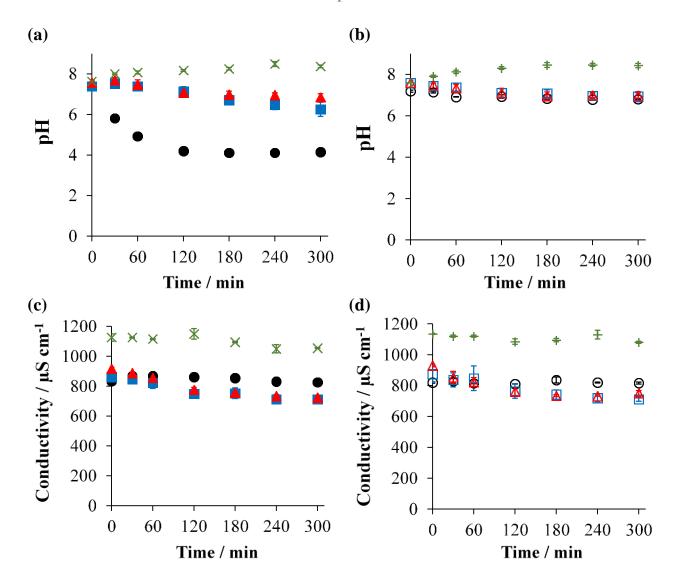


**Fig. IV.9**. Evolution of  $Mg^{2+}$  (**a**, **b**) and modeled interfacial  $CO_3^{2-}$  (**c**, **d**) concentrations during the electrolysis of electrolyte containing  $Ca^{2+}$  ( $\blacksquare$ ,  $\square$ ) or without  $Ca^{2+}$  ( $\times$ , +) at 4 mA cm<sup>-2</sup> (**a**, **c**) and 0.4 mA cm<sup>-2</sup> (**b**, **d**). Model curves of electrolyte in presence of  $Ca^{2+}$  ( $\longrightarrow$ ) or in absence of  $Ca^{2+}$  ( $\longrightarrow$ ). **Inset** (**a**): zoom of the initial kinetic evolution of  $Mg^{2+}$ . Anode: BDD, cathode: stainless steel and  $d_{elec}$ : 500  $\mu$ m.

#### IV.4.5. Evolution of pH and conductivity

In Fig. IV.10 are plotted the graphs of variation of bulk solution pH and ionic conductivity throughout the electrolysis of all investigated matrices of electrolyte at both high and low applied current densities. Few general trends can be drawn from these curves. Firstly, solution pH and ionic conductivity tended to decrease in the bulk during the electro-precipitation process. The decrease in pH was the most remarkable in the electrolyte without carbonates since no buffer effect of HCO<sub>3</sub><sup>-</sup>  $(CO_3^2)^2$  was present in this matrix. Similar remark was also made by other authors, Tlili et al. [47] and Piri et al. [26], when they checked upon the variation of pH in bulk solution during the formation of CaCO<sub>3</sub>. This drop of pH value was ascribed to the consumption of OH<sup>-</sup> that partake in the reaction of Mg(OH)<sub>2</sub> deposition according to Eq. (II.11). It could also be attributed to the pH participation towards the thermodynamic equilibrium shift of HCO<sub>3</sub><sup>-</sup>/CO<sub>3</sub><sup>2-</sup> in favor of CO<sub>3</sub><sup>2-</sup> formation in Eq. (I1.12), while consuming OH<sup>-</sup>. Net concentration of OH<sup>-</sup> in the bulk therefore declined, indicated by the decrease of measured bulk pH. The only case where bulk pH did not decrease was during the electrolysis of electrolyte in absence of Ca<sup>2+</sup>. At both applied current densities, CaCO<sub>3</sub> deposit was not formed. Meanwhile, OH was continuously electro-generated on the surface of cathode. OH diffusing from cathode surface to the bulk only reacted with existing HCO<sub>3</sub> until complete conversion into the basic form, i.e.  $CO_3^{2-}(Eq. (II.12))$ . Then, the accumulation of OH in the bulk yielded in an increase of pH value. Moreover, the only electro-precipitation that occurred in the electrolyte in absence of  $Ca^{2+}$  was  $Mg(OH)_2$  and it was only seen at higher range of  $j_{app}$ . However, as demonstrated in Fig. IV.10(a), OH consumed for the deposition of Mg(OH)<sub>2</sub> on cathode surface contributed trivially towards the diminishing pattern of pH in the bulk unlike during the deposition of CaCO<sub>3</sub>.

Similar argument can be done towards the trend of ionic conductivity in the bulk. Knowing the fact that OH<sup>-</sup> possesses high value of ionic conductivity despite not being the highest relatively to H<sup>+</sup>, its consumption throughout the electro-precipitating event of CaCO<sub>3</sub> manifested a decline in overall bulk ionic conductivity as shown in **Figs. IV.10(c)** and **IV.10(d)**.



**Fig. IV.10**. Evolution of pH (**a,b**) and ionic conductivity (**c,d**) during the electrolysis of different matrices of electrolyte (Ca<sup>2+</sup> and Mg<sup>2+</sup> without CO<sub>3</sub><sup>2-</sup> ( $\bigcirc$ , O), Ca<sup>2+</sup>, Mg<sup>2+</sup> and CO<sub>3</sub><sup>2-</sup> ( $\bigcirc$ ,  $\square$ ), Ca<sup>2+</sup> and CO<sub>3</sub><sup>2-</sup> without Mg<sup>2+</sup> ( $\triangle$ ,  $\triangle$ ), and Mg<sup>2+</sup> and CO<sub>3</sub><sup>2-</sup> without Ca<sup>2+</sup> ( $\times$ , +)) at 4 mA cm<sup>-2</sup> ( $\bigcirc$ ,  $\square$ ,  $\times$ ) and 0.4 mA cm<sup>-2</sup> ( $\bigcirc$ ,  $\square$ ,  $\triangle$ ). Anode: BDD, cathode: stainless steel,  $d_{elec}$ : 500 μm.

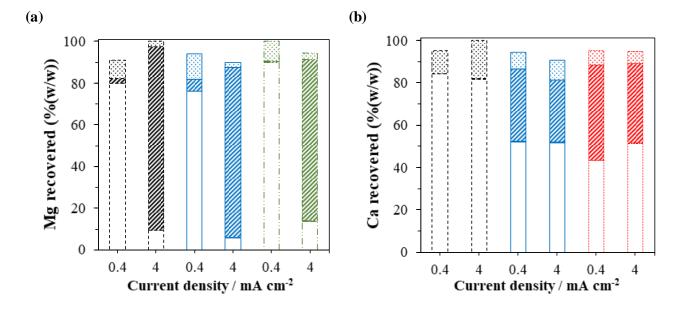
#### IV.4.6. Mass balance and elements recovery

**Figure IV.11** depicts the recovery of Mg and Ca from the different electrochemical reactor compartments at the end of experiment for different matrices of electrolyte at both current densities investigated. Very high percentage of Mg and Ca recovery across all experiments was verified with the lowest being 90%. The percentage of recovery across all electrolysis took an average value of 95%. From **Fig. IV.11**, in the case where electro-precipitation occurred, high percentage of Mg and Ca were recovered from cathode while negligible quantity was found in tubing and from anode surface. It is important to note that high Mg and Ca recovery means negligible loss of the elements was ensured. Mass balance of Mg and Ca from different compartment was conserved. Consequently, the trends of cation concentrations depicted in previous figures were reliable. Their decay from

electrolyte bulk, coupled to high mass recovery from cathode surface was indicative of mineral precipitations occurring on the cathode.

Looking at **Fig. IV.11(b)**, percentages of mass of Ca recovered from cathode in both matrices of electrolyte were higher at 0.4 mA cm<sup>-2</sup> as compared to 4 mA cm<sup>-2</sup>, confirming the relative influence of evolving gas activity on cathode surface. Moreover, the percentage of Ca recovery from cathode was always lower in the matrix in presence of Mg<sup>2+</sup>, whatever the current density applied. It is in agreement with the fact that less Ca<sup>2+</sup> was electro-precipitated in presence of Mg<sup>2+</sup> demonstrating the inhibition effect of the latter towards CaCO<sub>3</sub> co-precipitation with Mg(OH)<sub>2</sub>.

On the other hand, the recovery of Mg in all electrolytes under study illustrated in **Fig. IV.11(a)** confirmed that the  $Mg^{2+}$ -based deposit only took place under high current density. It further corroborated the fact that the formation of  $Mg(OH)_2$  deposit was not influenced by other ionic species co-present in the electrolyte.



**Fig. IV.11**. Percentages (w/w) of (a) Mg and (b) Ca recovered from different parts of experimental pilot ( $\square$ : electrolyte,  $\square$ : cathode,  $\square$ : anode and  $\square$ : tube) after 300 min-electrolysis at 0.4 and 4 mA cm<sup>-2</sup> of different matrices of electrolyte ((----): Ca<sup>2+</sup> and Mg<sup>2+</sup> without CO<sub>3</sub><sup>2-</sup>, (\_\_\_\_\_): Ca<sup>2+</sup>, Mg<sup>2+</sup> and CO<sub>3</sub><sup>2-</sup>, (\_\_\_\_): Ca<sup>2+</sup> and CO<sub>3</sub><sup>2-</sup> without Mg<sup>2+</sup>, and (----): Mg<sup>2+</sup> and CO<sub>3</sub><sup>2-</sup> without Ca<sup>2+</sup>).

# IV.4.7. Theoretical evolution of $Ca^{2+}$ , $Mg^{2+}$ and interfacial $CO_3^{2-}$ during electrolysis in different matrices

Simulations of evolution of Mg<sup>2+</sup> concentration during the electrolysis of all matrices of electrolyte depicted in Figs. IV.6(a), Fig. IV.6(b), Fig. IV.9(a) and Fig. IV.9(b) fitted very well with their corresponding experimental curves. Their RMSE values stayed underneath 0.0078 as summarized in Table IV.4 whilst ME and IOA values remained above 0.9849 and 0.9956 respectively under all case study. The modeling indicated that under the polarization of 4 mA cm<sup>-2</sup>, it took approximately 7 min for the electrochemical system to reach interfacial OH-crit (0.16 mol m<sup>-3</sup>, interfacial pH<sub>crit</sub> = 10.2) before Mg<sup>2+</sup> started to precipitate. This lag time was also observed experimentally as illustrated in inset of Fig. IV.6(a) and Fig. IV.9(a). The models of Mg<sup>2+</sup> concentrations decreased over time following the calibrated empirical value of  $k_{\rm Mg(OH)_2}$  of 0.330 m<sup>7</sup> mol<sup>-2</sup> min<sup>-1</sup>. Exactly same trend was observed across all matrices of electrolyte containing Mg<sup>2+</sup> since Mg(OH)<sub>2</sub> deposition was independent of CO<sub>3</sub><sup>2-</sup> and Ca<sup>2+</sup> ions. Accordingly, similar procedures were carried out for Ca(OH)<sub>2</sub> deposit. However, due to high solubility of  $Ca(OH)_2$ , the  $OH_{crit}^-$  concentration that needed to be reached was as high as 36.6 mol m<sup>-3</sup> (pH = 12.6). Under our investigated experimental setup, this  $OH_{crit}^-$  value was not achieved by either  $j_{app}$ . Maximum theoretical interfacial  $OH^-$  concentration was obtained under 4 mA cm<sup>-2</sup> at 7.43 mol m<sup>-3</sup> (pH = 11.8). As a consequence, no precipitation of Ca(OH)<sub>2</sub> was observed.

The modeling of interfacial  $CO_3^{2-}$  concentration at cathode surface was rather accurate as extensively discussed from Sections IV.4.4.1 to IV.4.4.3. In agreement with Eq. (IV.12) and Eq. (IV.15), interfacial  $CO_3^{2-}$  was proven to be dependent on  $j_{app}$ ; hence  $H_2$  gas evolution activity on cathode. Its concentration was as well influenced by the consumption of interfacial  $OH^-$ . Taking into account these matters, Eq. (IV.13) yields reliable models of evolution of  $Ca^{2+}$  concentration across multiple matrices of electrolyte investigated. The calibrated empirical value of constant rate of  $CaCO_3$  electroprecipitation ( $k_{CaCO_3}$ ) was  $3.6 \times 10^{-3}$  m<sup>4</sup> mol<sup>-1</sup> min<sup>-1</sup>.  $Ca^{2+}$  model curves fitted their experimental counterparts well. In agreement to listed values in Table IV.4, the lowest RMSE was 0.0761 in the system containing  $Mg^{2+}$ ,  $Ca^{2+}$  and  $CO_3^{2-}$  at low  $j_{app}$ . Other RMSE values ranged from 0.0828 to 0.1163 and the  $Ca^{2+}$  model fitting remained reliable in accordance to other model fitting criteria such as ME and IOA. Their respective values obtained for the case with worst RMSE fitting (i.e., matrix containing  $Ca^{2+}$  and  $CO_3^{2-}$  without  $Mg^{2+}$  at 0.4 mA cm<sup>-2</sup>) were 0.9758 and 0.9934.

In order to account for intensive concomitant  $H_2$  gas evolution at high  $j_{app}$ , the OH<sup>-</sup> hindrance factor  $\alpha$  introduced in **Eq. (IV.3)** and **Eq. (IV.15)** was taken at 7.5% to fit experimental results. This tells that approximately only 7.5% of cathode surface was effective for mineral electro-precipitation

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including the detachment of deposits already formed by the evolving gas. The  $\alpha$  value was found to approximately match the observed experimental decrease of CaCO<sub>3</sub> scaling (7.2%) between the two regions of applied current densities investigated. In addition, it has also been observed that at lower applied current, the applied current density already surpassed limiting current density (Section IV.4.2). This means that secondary reaction like concurrent gas evolution was already present but it did not influence on the availability of interfacial  $CO_3^{2-}$  on cathode (Fig. IV.8(b)) but rather improving the mass transfer of  $Ca^{2+}$  and  $Mg^{2+}$  towards cathode surface (Section IV.4.4.1). The hindrance of  $CaCO_3$  growth kinetics in this case was purely attributed to steric inhibition of  $Mg^{2+}$  in solution since no  $Mg(OH)_2$  was produced under this condition [64]. Therefore, the kinetic law of depositing  $CaCO_3$  was slightly modified taking into account this effect as defined in Eq. (IV.21):

$$r_{Ca^{2+}} = -r_{CaCO_3} = -(k_{CaCO_3} - k_{inhib}) [Ca^{2+}] [CO_3^{2-}]_{int,t}$$
 (IV.21)

where  $k_{\rm inhib}$  (in m<sup>4</sup> mol<sup>-1</sup> min<sup>-1</sup>) is the steric inhibition of Mg<sup>2+</sup> ion either by adsorption or incorporation into CaCO<sub>3</sub> lattices as previously discussed in Section IV.4.4.2. By fitting the modeled Ca<sup>2+</sup> concentration with its experimental data (the matrix in presence of Mg<sup>2+</sup>) at low  $j_{\rm app}$ ,  $k_{\rm inhib}$  was evaluated to be  $9.0 \times 10^{-4}$  m<sup>3</sup> mol<sup>-1</sup> min<sup>-1</sup>. This value is about 25% of the empirical CaCO<sub>3</sub> electroprecipitation constant of  $k_{\rm CaCO_3}$ . It suggests that in the presence of 5 mg L<sup>-1</sup> of Mg<sup>2+</sup> (in the system containing 150 mg L<sup>-1</sup> of Ca<sup>2+</sup>), 25% less CaCO<sub>3</sub> electro-precipitation should have been observed due to the inhibition effect. However, experimental results indicated only 12% less CaCO<sub>3</sub> scaling after 5 h of electrolysis, which is about half the theoretical value. This discrepancy was attributed to the enhancement of Ca<sup>2+</sup>, CO<sub>3</sub><sup>2-</sup> and Mg<sup>2+</sup> transport by micro- and macro-convection towards cathode interface [56-59] which could favor the electro-precipitation reactions.

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**Table IV.4**. Model fitting evaluation between simulated and experimental evolution of Mg<sup>2+</sup> and Ca<sup>2+</sup> concentrations in all investigated matrices of electrolyte, considering RMSE, ME and IOA at both 0.4 and 4 mA cm<sup>-2</sup>. (N/A: division by 0 values. Appeared when evaluating the model in the case where no deposition occurs).

Electrical and a	T	Model fitting	Applied current density		
Electrolyte	Ionic species	criteria	0.4 mA cm <sup>-2</sup>	4 mA cm <sup>-2</sup>	
	$\mathrm{Mg}^{2+}$	RMSE	0.0073	0.0064	
		ME	N/A	0.9901	
$Ca^{2+}$ and $Mg^{2+}$		IOA	N/A	0.9975	
without CO <sub>3</sub> <sup>2-</sup>		RMSE	0.0999	0.1036	
	$Ca^{2+}$	ME	N/A	N/A	
		IOA	N/A	N/A	
G 21 1 GO 2		RMSE	0.1163	0.1136	
$Ca^{2+}$ and $CO_3^{2-}$ without $Mg^{2+}$	Ca <sup>2+</sup>	ME	0.9758	0.9710	
without Mg		IOA	0.9934	0.9918	
Mg <sup>2+</sup> and CO <sub>3</sub> <sup>2-</sup> without Ca <sup>2+</sup>	$\mathrm{Mg}^{2+}$	RMSE	0.0034	0.0078	
		ME	N/A	0.9849	
		IOA	N/A	0.9956	
$Ca^{2+}$ , $Mg^{2+}$ and $CO_3^{2-}$	${ m Mg^{2+}}$	RMSE	0.0056	0.0060	
		ME	N/A	0.9883	
		IOA	N/A	0.9969	
		RMSE	0.0761	0.0828	
	$Ca^{2+}$	ME	0.9844	0.9767	
		IOA	0.9960	0.9938	

#### IV.5. Conclusions

The occurrence of mineral electro-precipitation inside a microfluidic reactor was newly investigated in this work. Due to the existence of precipitating agents in water resources of natural or anthropogenic origin, electro-precipitates can be formed during advanced electro-oxidation processes and they might compromise the efficiency of such processes. Inside a scalable microfluidic reactor, the phenomenon of local alkalization on cathode surface originated principally from reduction of water and not dissolved O<sub>2</sub> under investigated conditions. The reaction kinetics at the proximity of cathode surface were limited by mass transfer. Upon further investigation, the influence of electromigration towards the occurrence of electro-precipitation was negligible, meaning that diffusion was the main limiting factor.

MgCO<sub>3</sub> deposit was not observed throughout this work, while Mg(OH)<sub>2</sub> precipitate was highly dependent on  $j_{app}$ , because the local pH value was the decisive parameter upon its formation. Thus Mg(OH)<sub>2</sub> was hardly formed at low  $j_{app}$ . In addition, Mg(OH)<sub>2</sub> deposition was independent of ionic species jointly present in the electrolyte bulk. Regarding Ca-based precipitates, Ca(OH)<sub>2</sub> was not formed under our investigated configuration, while CaCO<sub>3</sub> was easily deposited on cathode surface even under small current polarization. More electro-precipitation of CaCO<sub>3</sub> was recovered at lower  $j_{app}$  (0.4 mA cm<sup>-2</sup>) owing to vigorous gas evolution at higher  $j_{app}$  (4 mA cm<sup>-2</sup>). CaCO<sub>3</sub> scaling was proven to be impeded by the presence of Mg<sup>2+</sup> ion. A novel mathematical model was proposed to simulate and predict the evolution of depositing species of Mg<sup>2+</sup>, Ca<sup>2+</sup> and interfacial CO<sub>3</sub><sup>2-</sup> in presence or absence of co-deposition within the microfluidic reactor. The models fitted well with experimental data given the RMSE, ME and IOA values.

It could be deduced eventually that  ${}^{\bullet}$ OH continuously produced on BDD anode did not influence towards the electro-precipitation occurring on cathode counterpart in the applied condition despite micrometric distance separating them. In order to minimize scaling formation, a compromise with respect to degradation efficiency of pollutant needs to be accommodated for upcoming research. Lower  $j_{\rm app}$  could be used, but lower degradation of pollutant via direct electron transfer or indirect oxidation would be consequently expected due to lower overpotential. Alternatively, electro-precipitation could be minimized by applying high current density, benefiting the intensified evolving gas on cathode surface to un-do the deposits. However, energy consumption would be deemed deal-breaker for its application at industrial scale. Nonetheless, since micrometric range of  $d_{\rm elec}$  extensively cut down the internal resistance of electrochemical cells, cell voltage would not be eventually so high at such high current densities. Further studies are required to test microfluidic configuration in order

to concomitantly yield high degradation efficiency while minimizing the resulting mineral electroprecipitation.

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# V. Mass transfer evolution in microfluidic thin film electrochemical reactor: New correlations from millimetric to submillimetric interelectrode distances

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#### V.1. Introduction

Electrochemical reactor with micrometric feature has brought interesting approach in multiple disciplines such as electrosynthesis [1-5], electrodeposition [6, 7], sensor development [8, 9] and wastewater treatments [10-15]. Versatility, small footprint, suitability for fast kinetic reactions, easily thermoregulated, independence of supporting electrolyte, mass transfer intensification and huge energy saver are amongst the leverages that are sought from the microfluidic flow reactor technology [16-18]. Mass transfer characterization in submillimetric range has already been scrutinized since the early 70's with the development of flow rectangular channel, tubular and granular bed electrochemical reactors [19-27]. Estimating the average  $k_{\rm m}$  is particularly crucial for sizing the electrochemical reactors and for scale-up studies [28]. Regarding millimetric scale reactors, mass transfer correlations of different geometries with or without turbulence promoters [28-33], electrodes morphology [34, 35] as well as innovative electrolyte inlet [36, 37] and agitation system [38, 39] have been widely investigated in literature. However, more data are required on the mass transfer evolution and correlation in microfluidic parallel-plate reactors which are applicable at larger scale [40]. It could be due to challenging experimentation in different hydrodynamics conditions and temperature as well as to the difficulty to configure 3-electrode setups under narrow gap [38, 39]. Hence, it has been reported for the first time here-in the mass transfer evolution over a wide range of  $d_{\rm elec}$  from millimetric (3 mm) down to submillimetric value (100 µm) in a flow-by parallel-plate reactor with a large effective surface area. Its design is more apt for scale-up application in industries. A model has been proposed to evaluate  $k_{\rm m}$  as function of  $d_{\rm elec}$ . Finally, mass transfer correlation involving dimensionless numbers in microfluidic and millimetric feature has been proposed.

#### V.2. Experimental section

#### V.2.1. Electrochemical system

The electrochemical setup consisted of a parallel-plate reactor and its exploded view is illustrated in **Fig. III.2** in **Chapter 3**. Both working and counter electrodes were planar rectangles, facing one another. Stainless steel was the WE while BDD served as CE. Their W and L were 5 and 10 cm respectively giving out an effective surface area of 50 cm<sup>2</sup>. A spacer made of PTFE with varying

thicknesses of 100, 250, 500, 1000, 1500, 2000 and 3000  $\mu$ m was used to separate the WE and CE. Thus, they defined the varying  $d_{elec}$  investigated in this work, distinguished from micrometric progressively into a typical millimetric configuration. Ag-AgCl was used as the RE.

#### V.2.2. Mass transfer characterization

The characterization of mass transfer in the electrochemical cells was carried out following the widely adopted limiting current technique method as described elsewhere [41, 42]. 0.5 L of electrolytic solution containing 0.05 M K<sub>3</sub>Fe(CN)<sub>6</sub>, 0.10 M K<sub>4</sub>Fe(CN)<sub>6</sub> and 0.50 M Na<sub>2</sub>CO<sub>3</sub> as supporting electrolyte was employed. High concentration of supporting electrolyte was used to avoid electromigration effect [41, 43]. Using this technique, the reduction of ferricyanide  $(Fe(CN)_6^{3-})$  was brought to a maximum rate (limiting current plateau) so that its reduction was purely controlled by mass transfer regime. The physical properties of electrolyte at different investigated temperatures [29, 30, 44] are given in Table V.1. LSV was swept in the region of cathodic potential activity of ferriferrocyanide electrolyte over a range of potentials from 0.2 to -1.6 V versus Ag-AgCl reference electrode at 2.5 mV s<sup>-1</sup> scan rate. Within the adopted range of flow rates, the scan rate was sufficiently slow to attain steady-state [45]. The BDD CE features a wide potential window of water stability with O<sub>2</sub> evolution reaction (OER) occurring at 2.5-2.7 V/SHE. At this high OER potential, negligible interference of O<sub>2</sub> evolved on the CE was expected in the range of investigated cathodic potential. Moreover, to further confirm the influence of the presence of dissolved O2 in the electrolyte, preliminary experiments were conducted at both extremities (i.e., 100 and 3000 µm) as well as using 500 µm interelectrode gap under the condition free of dissolved O2. The electrolyte in the reservoir was aerated with N<sub>2</sub> gas for 30 min before LSV was performed. The results indicated that the limiting current plateau (just before the reduction of solvent) were not influenced by the presence of dissolved  $O_2$ . The outcoming limiting current values ( $I_{lim}$  in A) under  $d_{elec}$  and investigated operating conditions were related to average  $k_{\rm m}$  (m s<sup>-1</sup>) via **Eq. (V.1) [42]**:

$$k_{\rm m} = \frac{I_{\rm lim}}{n_{\rm e} \, F \, S \, C_{\rm SOL}} \tag{V.1}$$

where  $n_e$  is the number of electrons involved in the reduction reaction of ferri- to ferrocyanide (i.e., 1), F the Faraday's constant (96485 C mol<sup>-1</sup>), S the effective surface area of working electrode (m<sup>2</sup>) and  $C_{SOL}$  the molar concentration of ferricyanide in the bulk (mol m<sup>-3</sup>).

During the characterization of  $k_{\rm m}$  over the range of  $d_{\rm elec}$ , electrolyte cross-sectional velocity,  $u_{\rm L}$ , in m min<sup>-1</sup> (**Eq. (V.2)**) inside electrochemical reactors was kept constant (4 m min<sup>-1</sup>) from one distance to another.

$$u_{\rm L} = \frac{\dot{V}_{\rm F}}{A_{\rm C}} \tag{V.2}$$

$$A_{\rm C} = d_{\rm elec} W \tag{V.3}$$

where  $\dot{V}_{\rm F}$  is the volumetric flow rate (in m<sup>3</sup> min<sup>-1</sup>), W is the electrode width (in m),  $A_{\rm C}$  is the cross-sectional area of electrolyte passage (in m<sup>2</sup>) and  $d_{\rm elec}$  is the interelectrode distance (in m).

This was to investigate the influence of  $d_{\rm elec}$  on  $k_{\rm m}$ , without considering the enhancement of mass transfer related to convective flow of electrolyte. To achieve the purpose, the electrolyte flow rate was adjusted accordingly to the volume of reactor. Consequently, not only the superficial velocity was identical, but electrolyte residence time ( $\tau$ ) in min (Eq. (V.4)) was also kept identical (0.025 min) from one configuration to another.

$$\tau = \frac{\dot{V}_{\rm F}}{V_{\rm R}} \tag{V.4}$$

$$V_{\rm R} = d_{\rm elec}WL \tag{V.5}$$

where  $V_R$  is the cell volume (in m<sup>3</sup>) and L is the electrode length (in m).

The flow rate varied from 10 to 600 mL min<sup>-1</sup>, when  $d_{\rm elec}$  increased from 100 to 3000  $\mu$ m. Across all investigated  $d_{\rm elec}$  and for relevant comparison, electrolyte flow was kept under laminar regime according to the following range of Reynolds (Re) values: 7 to 623. Owing to the entrance manifolds of tube-type entry of the electrochemical cell, under similar hydrodynamic regimes, electrolyte recirculating zone in the reactor was considered minimal in all investigated configurations [36, 46, 47].

**Table V.1**. Physical properties of ferri-ferrocyanide electrolytic solution at 10, 25 and 50 °C.

Temperature / °C	Density, $ ho_{ m L}$ / kg m <sup>-3</sup>	Kinematic viscosity, $\nu_L  /  m^2  s^{\text{-}1}$	Diffusion coefficient, $D_{\rm L}$ / ${\rm m}^2~{\rm s}^{-1}$
10	1042.06	$14.28 \times 10^{-7}$	$4.31 \times 10^{-10}$
25	1038.04	$9.82 \times 10^{-7}$	$6.48 \times 10^{-10}$
50	1028.87	$6.06 \times 10^{-7}$	$11.39 \times 10^{-10}$

#### V.3. Results and discussion

#### V.3.1. Mass transfer behavior over a wide range of interelectrode distances

Voltammograms of electrochemical reactors with different  $d_{\rm elec}$  study were illustrated in Fig. V.1. From the plateau of  $I_{\rm lim}$  identified on the voltammograms, average  $k_{\rm m}$  values over the wide range of  $d_{\rm elec}$  are shown in Fig. V.2. Interestingly, two distinct behaviors of  $k_{\rm m}$  were observed as a first approach. The increase of  $k_{\rm m}$  in the millimetric range from 3000  $\mu$ m down to 1000  $\mu$ m seemed linear (slope:  $-10^{-3}$  s<sup>-1</sup>, coefficient of determination  $R^2 = 0.91$ ). Beneath 1000  $\mu$ m,  $k_{\rm m}$  increased drastically with decreasing  $d_{\rm elec}$  further down into the micrometric ranges. This trend demonstrates that the mass transfer was massively enhanced while transitioning from millimetric to micrometric gap, which confirms the interest of microfluidic reactors for transfer intensification [14, 16, 48-53]. The range of  $k_{\rm m}$  determined under both millimetric and micrometric configurations in this work are in close agreement with those reported in literature (Table II.3 in Chapter 2). In millimetric ranges,  $k_{\rm m}$  values of  $1.61 \times 10^{-5}$  m s<sup>-1</sup> and  $1.82 \times 10^{-5}$  m s<sup>-1</sup> were obtained respectively for 3000  $\mu$ m and 1000  $\mu$ m, in comparison to reported values of  $k_{\rm m}$  between  $0.30 - 1.20 \times 10^{-5}$  m s<sup>-1</sup> using 5000  $\mu$ m [54] and  $1.60 \times 10^{-5}$  m s<sup>-1</sup> using 1000  $\mu$ m [28].

Levich equation (**Eq.** (**V.6**)) has been widely adopted in literature to relate  $I_{\text{lim}}$  to electrolyte volume flow rate in thin rectangular channel electrode reactors [19, 55, 56].

$$I_{\text{lim}} = 0.925 n_{\text{e}} FW C_{\text{SOL}} (D_{\text{L}} x_{\text{e}})^{2/3} \left(\frac{\dot{V}_{\text{F}}}{h^2 d}\right)^{1/3}$$
 (V.6)

where

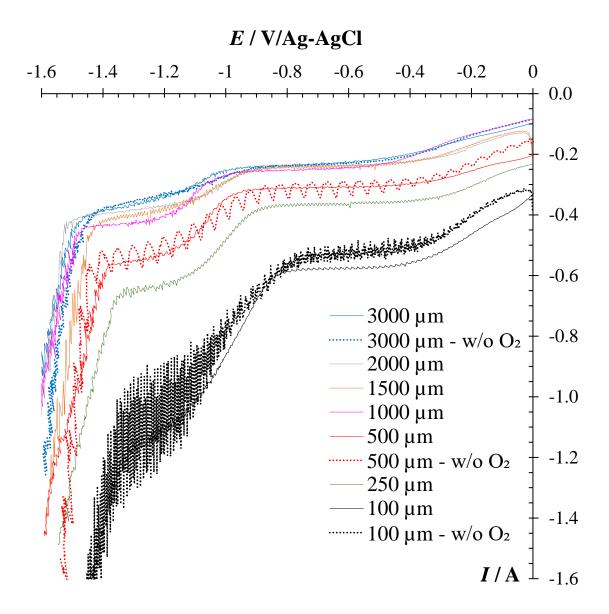
$$\dot{V}_{\rm F} = u_{\rm L} W d_{\rm elec} \tag{V.7}$$

$$h = \frac{d_{\text{elec}}}{2} \tag{V.8}$$

W is the electrode width (in cm),  $x_e$  is the electrode length (equivalent to L, in cm),  $\dot{V}_F$  in cm<sup>3</sup> s<sup>-1</sup>, h is half the cell depth (in cm), d is the channel width (equivalent to W in this work, in cm) and  $u_L$  in cm s<sup>-1</sup>. **Equation (V.6)** is ideally applicable under conditions where  $h \ll d$ , sufficiently large electrode area, the electrolyte flow rate is not too low and the profile of electrolyte flow velocity in the channel is parabolic [19]. Relating **Eq. (V.1)** and **Eq. (V.6)** gives out an expression of average  $k_m$  in function of  $d_{\text{elec}}$  (**Eq. (V.9**)):

$$k_{\rm m} = \frac{2.94 \, D_{\rm L}^{2/3} \, u_{\rm L}^{1/3} \, W \, L^{2/3}}{S \, d_{\rm elgc}^{1/3}} \tag{V.9}$$

Slight adjustment of **Eq.** (**V.6**) coefficient (2.94) was done to match the experimental data and this deviation could be due to the non-ideality of direct application of **Eq.** (**V.6**) under the condition where electrode width and channel width are identical (W = d), i.e. in which an edge effect could not be neglected [19, 40]. The plot of experimental and theoretical  $k_{\rm m}$  versus  $d_{\rm elec}$  is given in **Fig. V.2**. The theoretical curve matches well with the experimental counterpart with a deviation of 0.000187 when they were subjected to RMSE (**Eq.** (III.3)).



**Fig. V.1**. LSV of the electrochemical cell with all investigated  $d_{\rm elec}$ . WE: stainless steel, CE: BDD and RE: Ag/AgCl. 0.5 L of electrolyte consisted of 0.05 M K<sub>3</sub>Fe(CN)<sub>6</sub>, 0.10 M K<sub>4</sub>Fe(CN)<sub>6</sub> and 0.50 M Na<sub>2</sub>CO<sub>3</sub>. Cathodic potential ranges from 0.2 to -1.6 V vs. Ag-AgCl at 2.5 mV s<sup>-1</sup> scan rate. Dotted line: experiments in the absence of dissolved O<sub>2</sub> by purging with N<sub>2</sub> gas in the feed tank.

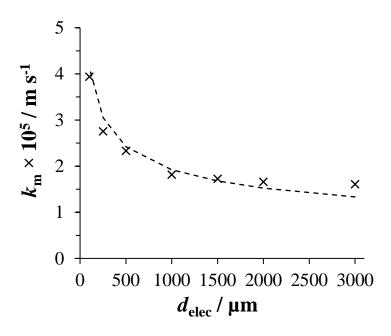


Fig. V.2. Variation of experimental and theoretical average  $k_{\rm m}$  as function of  $d_{\rm elec}$ . Operating conditions: Flow rate:  $10-600~{\rm mL~min^{-1}}$ , temperature:  $25~{\rm ^{\circ}C}$ ,  $u_{\rm L}$ : 4 m min<sup>-1</sup> and  $\tau$ : 0.025 min.

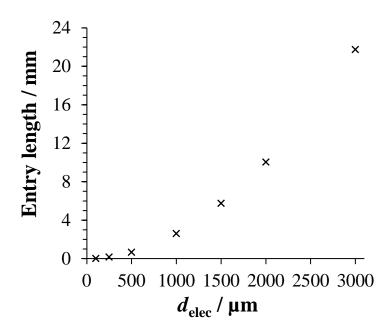
### V.3.2. Mass transfer correlations in microfluidic and millimetric parallel-plate electrochemical reactors

Evaluation of mass transfer correlations were conducted on electrochemical reactors with 100 and 500  $\mu$ m (representing micrometric setup) and 3000  $\mu$ m (representing millimetric setup). The variation of mass transfer with fluid hydrodynamics, fluid physicochemical properties and eventually electrochemical reactor geometry can be expressed using theoretical dimensionless Sherwood (*Sh*) correlation accounting for *Re*, Schmidt (*Sc*), dimensionless distance (*Le*) numbers as well as  $d_{\rm elec}/W$  aspect ratio written in Eq. (V.10) [57].

$$Sh = aRe^b Sc^c Le^d \left(\frac{2}{1 + \frac{d_{\text{elec}}}{W}}\right)^{1/3} \tag{V.10}$$

The form of expression in **Eq.** (**V.10**) is valid with strict respect to the Lévêque approximation [55, 58], which imposes that (i) the electrolyte is incompressible having constant physical properties, (ii) the flow is in laminar steady-state, becomes fully developed after a certain entry length for a given reactor geometry, (iii) mass transfer is due to convection and diffusion, (iv) electrolyte flow is treated as bidimensional in x-y planes (**Fig. III.2**), (v) molecular diffusion of reactive species only occurs in x-axis (normal to the electrodes), (vi) friction at walls induced retardation and produces parabolic velocity profile with maximum velocity at the center of the channel and (vii) diffusional double-layer is a lot smaller than hydrodynamic boundary layer. This dimensionless expression is particularly useful to estimate the average  $k_{\rm m}$  values during scale-up or scale-down of an electrochemical reactor

at the operating conditions of interest. The entry length ( $E_L$ ) was evaluated via **Eq. (V.11)** [58, 59] and they were plotted in **Fig. V.3**. It increased from 0.027 to 21.7 mm when  $d_{elec}$  increased from 100 to 3000  $\mu$ m.



**Fig. V.3**. Variation of entry length of electrochemical cell with different  $d_{\rm elec}$ .

Re, Sc and Le are defined by Eqs. (V.12)-(V.14) whilst a, b, c and d are constant parameters.

$$E_{\rm L} = 0.01 d_{\rm H} Re \tag{V.11}$$

$$Re = \frac{u_{\rm L} d_{\rm H}}{v_{\rm L}} \tag{V.12}$$

$$Sc = \frac{v_{\rm L}}{D_{\rm L}} \tag{V.13}$$

$$Le = \frac{d_{\rm H}}{L} \tag{V.14}$$

$$d_{\rm H} = \frac{2 W d_{\rm elec}}{W + d_{\rm elec}} \tag{V.15}$$

 $D_{\rm L}$  and  $v_{\rm L}$  are diffusion coefficient and kinematic viscosity respectively both in m<sup>2</sup> s<sup>-1</sup> taken at the operated temperature [29, 30, 44] (Table V.1).  $d_{\rm H}$  is hydraulic diameter in m.

To establish the theoretical Sh correlation,  $k_{\rm m}$  values were measured at different electrolytic flow rates and temperatures. Experimental Sh were computed using Eq. (V.16) [42] and they were plotted against varying investigated Re and Sc numbers. The coefficients a, b, c and d were extracted from the best fitting between theoretical and experimental Sh plots.

$$Sh = \frac{k_{\rm m} d_{\rm H}}{D_{\rm L}} \tag{V.16}$$

The influence of electrolyte flow rate and temperature on  $k_{\rm m}$  for each selected configuration is presented in **Fig. V.4**.  $k_{\rm m}$  increased with increasing flow rate across both configurations at constant temperature (**Fig. V.4(a**)). This finding is expected owing to the enhancement of mass transfer brought to the  $Fe(CN)_6^{3-}$  reduction reaction, when higher flow rate was tested. It can be noticed that the  $k_{\rm m}$  increase with increasing electrolyte flow rate was markedly pronounced in microfluidic device (100 and 500 µm) in comparison to the distance of 3000 µm (**Fig. V.4(a**)). The highest  $k_{\rm m}$  (5.06 × 10<sup>-5</sup> m s<sup>-1</sup> at 0.05 L min<sup>-1</sup>) was achieved when the electrochemical reactor was used with 100 µm interelectrode gap. In addition, when the temperature varied from 10 to 50°C,  $k_{\rm m}$  increased from 3.28 × 10<sup>-5</sup> to 5.22 × 10<sup>-5</sup> m s<sup>-1</sup> for the 100 µm, from 1.66 × 10<sup>-5</sup> to 3.11 × 10<sup>-5</sup> m s<sup>-1</sup> for the 500 µm and from 1.20 × 10<sup>-5</sup> to 2.41 × 10<sup>-5</sup> m s<sup>-1</sup> for the 3000 µm gap (**Fig. V.4(b**)). These trends were due to the increase in diffusivity of  $Fe(CN)_6^{3-}$  reducing species with temperature [44, 60, 61].

Using the  $k_{\rm m}$  values obtained under different operating conditions, experimental Sh were evaluated as illustrated in Fig. V.5. The best fitting of Eq. (10) to the experimental data of each submillimetric and millimetric configuration were achieved when the coefficients a, b, c and d took the values provided in Table V.2. The coefficient a slightly varied: it decreased from 3.00 to 2.31 from 100  $\mu$ m to 500  $\mu$ m  $d_{\rm elec}$  before rising up again to 3.01 using 3000  $\mu$ m configuration. The values were within the range reported in literature [57, 62]. b variation between 0.32 and 0.37 corroborated the reported Re exponent value close to 0.30 in literature that corresponds to fully developed laminar flow [34, 37, 57, 62, 63]. The widely reported 1/3 ratio of Sc exponent c as well as of Le exponent d were obtained to match the experimental data under all three setups [30, 32, 34, 37, 57, 62, 63].

The fitting of theoretical Sh correlations with experimental data for all three  $d_{\rm elec}$  configurations were evaluated using RMSE (Eq. (III.3)). Deviation of 0.038, 0.056 and 0.024 were calculated between logarithmic plots of theoretical-experimental Sh against Re for 100, 500 and 3000  $\mu$ m configurations respectively. In function of Sc, theoretical plot deviated from the experimental data by 0.070, 0.047 and 0.026 in respective order of 100, 500 and 3000  $\mu$ m. Therefore, theoretical Sh correlations fitted very well with their experimental counterparts.

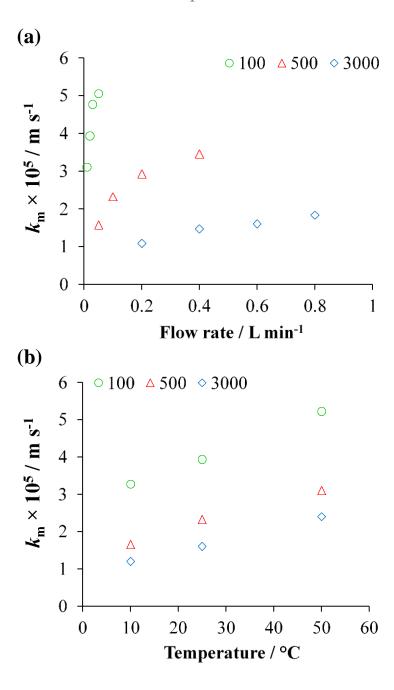
Given the trend of a, b, c and d coefficients under the large spectrum of investigated  $d_{\text{elec}}$ , under fully developed laminar regime, an average Sh correlation written in Eq. (V.17) can be proposed.

$$Sh_{\text{ave}} = 2.7733Re^{0.35}Sc^{1/3}Le^{1/3}\left(\frac{2}{1+\frac{d_{\text{elec}}}{W}}\right)^{1/3}$$
 (V.17)

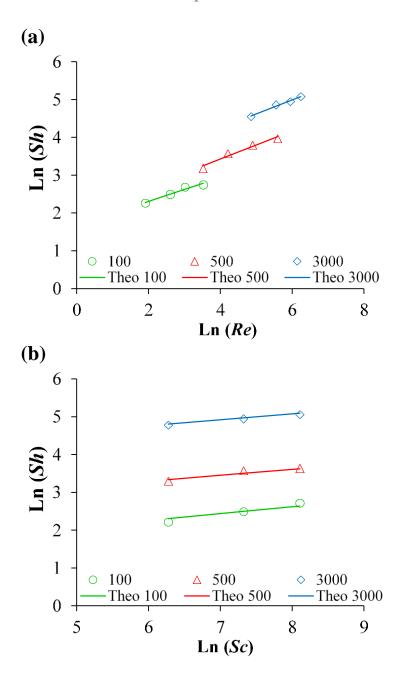
The average Sh correlation is in close agreement with the one proposed by Pickett et al. [57] with the a coefficient slightly higher in this present work. It could be attributed to a larger surface area used here (50 cm<sup>2</sup> vs. 16 cm<sup>2</sup> in [57]) as well as enhancement of mass transfer brought to the microfluidic reactor.

**Table V.2.** Fitted values of a, b, c and d of theoretical Sh correlation in parallel-plate electrochemical reactors under microfluidic and millimetric configurations under laminar regime.

Configuration	а	b	c	d	Re range	Sc range
Thin film reactor	3.00	0.32	0.33	0.33	7 - 34	532 - 3315
(distance: 100 µm)	3.00	0.32	0.33	0.55	7 - 34	332 - 3313
Thin film reactor	2.31	0.37	0.33	0.33	34 - 269	532 - 3315
(distance: 500 µm)	2.31	0.37	0.55	0.55	34 - 207	332 - 3313
Parallel-plate	3.01	0.37	0.33	0.33	128 - 623	532 - 3315
(distance: 3000 µm)	3.01	0.57	0.55	0.55	33 120 - 023	332 - 3313



**Fig. V.4**.  $k_{\rm m}$  as function of (a) electrolyte flow rate and (b) temperature for  $d_{\rm elec}$  of 100, 500 and 3000  $\mu$ m. **Operating conditions**: (a) Temperature = 25 °C and (b) flow rate = 0.02, 0.10 and 0.60 L min<sup>-1</sup> for  $d_{\rm elec}$  of 100, 500 and 3000  $\mu$ m respectively.



**Fig. V.5**. Experimental (symbol) versus theoretical *Sh* correlation (continuous line) as function of (a) *Re* and (b) *Sc* numbers. **Operating conditions**: (a) Temperature = 25 °C and (b) flow rate = 0.02, 0.10 and 0.60 L min<sup>-1</sup> for  $d_{\rm elec}$  of 100, 500 and 3000  $\mu$ m respectively.

#### V.4. Conclusions

The variation of  $k_{\rm m}$  as function of  $d_{\rm elec}$  from millimetric to micrometric range has been newly reported. A novel single equation to estimate  $k_{\rm m}$  over a wide range of  $d_{\rm elec}$  has been proposed. It highlights the inverse relationship between  $k_{\rm m}$  and  $d_{\rm elec}$ , which makes  $k_{\rm m}$  values greatly increased at distances below 1000  $\mu$ m. Using experimental  $k_{\rm m}$  data in submillimetric as well as millimetric configurations at varying flow rates (7 < Re < 623) and temperatures (532 < Sc < 3315), mass transfer correlations taking into consideration not only fluid properties but also electrochemical reactor hydrodynamic and geometry, have been established. The theoretical Sh number plots matched very well with their corresponding experimental values. The newly established Sh correlation in this work covered a wide interval of interelectrode distances. They were obtained using an electrochemical cell whose design is more adapted for multidisciplinary applications at larger scale whilst the robustness of the Sh correlation was experimentally justified. The results presented here-in will be particularly helpful for  $k_{\rm m}$  estimation in a parallel-plate reactor under wide range of  $d_{\rm elec}$  (i.e., from several tens of micrometers to several millimeters) at given applied operating conditions. These expressions can be particularly suitable for micro-distances that can be useful for great mass transfer intensification, unlike millimetric or even centimetric distances often implemented.

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# VI. Role of interelectrode distance and electrogenerated gas bubbles on mineral electro-precipitation

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#### VI.1. Introduction

The application of microfluidic reactors within the framework of EAOPs has offered another viewpoint in terms of wastewater treatment capability [1-4]. Coupled to the development of diamond-based anode or other high overpotential for OER, higher degree of electrochemical oxidation of organic contaminants has been reached [5-8]. The micrometric  $d_{\text{elec}}$  not only allow mineralization of organic pollutants without the addition of supporting electrolyte but it also vastly accommodates the phenomenon of limitation by mass transfer and ohmic resistance [9]. This is particularly adequate with solutions having low ionic conductivity, which is typical of the effluent at outlet of municipal WWTPs [10-12].

Numerous studies have already been reported on the formation of mineral scaling in electrochemical cells while treating different water sources originated from different domains of application: artificial water for cooling system in cooling towers and oil refineries [13, 14], artificial [15, 16] and natural underground water [17, 18], mineral potable water [19, 20], artificial [21-24] and real seawater [25, 26]. Most of the works were carried out using rotating disk electrodes (RDE) possessing effective surface area of 0.2 cm<sup>2</sup> [15-19, 21, 22, 25], 0.5 cm<sup>2</sup> [13, 14, 27], 1 cm<sup>2</sup> [24, 28, 29] and 1.5 cm<sup>2</sup> [23]. Recently, more scalable electrochemical cell geometries were suggested to study the occurrence of mineral electro-precipitation. Artificial brine solution (25 mL) was used to precipitate Ca and Mg deposits on cathode with 0.75 cm<sup>2</sup> effective surface area in a parallel-plate electrochemical cell working in discontinuous mode [30, 31]. The cathode and anode were separated by a 5 mm gap. In another series of work, horizontal parallel-plate cell with higher geometric surface area (i.e., 36 cm<sup>2</sup>) was used to electrochemically precipitate calcium, magnesium and phosphorus [32-35]. 0.8 to 1 L wastewater effluent was electrolyzed in recirculation mode and the distance separating anode and cathode in these works was 30 mm. Elsewhere, Sanjuán et al. in their work conducted a water hardness treatment of simulated electrodialysis concentrate [36]. The studies were done in a pilot plant working in flow mode using commercial undivided cell with 100 cm<sup>2</sup> surface area electrodes. Amongst the list, our recent work [37] was the first to illustrate the occurrence of mineral scaling inside a parallel-plate microfluidic reactor with 50 cm $^2$  surface area and 500  $\mu$ m  $d_{\rm elec}$ . The influence of electrolytic composition and current density were scrutinized. To the best of authors' knowledge, no report has yet been published to discuss the variation of mineral deposition with  $d_{\rm elec}$  lower than 5

mm, especially at submillimetric distances that imply a different mass transfer behavior (Chapter 4). Moreover, it is known that electro-precipitation often takes place with concomitant HER [28, 29, 38-40]. It results in calcareous deposit with porous morphology and is also the reason for partial deposit detachment from cathode surface [27, 29, 39, 41]. However, the role of gas evolution on electro-precipitation in submillimetric distances has never been explored. Furthermore, mass transfer intensification would promote the involvement of O<sub>2</sub> gas generated from OER at anode under micrometric configuration.

For the first time, the influence of a wide range of  $d_{\rm elec}$  (50-3000 µm) with regard to the occurrence of cathodic mineral deposition was therefore investigated in details. The involvement of HER while varying  $d_{\rm elec}$  and its effect towards the quantity of mineral scaling were further scrutinized. A new kinetic model has been proposed to predict the electro-precipitation at the different  $d_{\rm elec}$ , by incorporating the gas evolution. To reinforce the mechanistic understanding at the cathode/electrolyte interface over the wide range of investigated  $d_{\rm elec}$ , EIS analysis were performed.

#### VI.2. Experimental section

Synthetic effluent contained fixed concentrations of  $Ca^{2+}$  (150 mg  $L^{-1}$ ),  $Mg^{2+}$  (5 mg  $L^{-1}$ ) and 60 mg-C  $L^{-1}$  carbonates, which were dissolved in ultrapure water. The initial value of pH was adjusted to 7.6 by the addition of sulfuric acid when necessary. The mixture of salts including pH adjustment yielded an ionic conductivity of around 1000  $\mu$ S cm<sup>-1</sup> and ionic strength of 0.02 mol  $L^{-1}$ . These properties correspond to the average characteristic of effluent at the outlet of municipal WWTPs. BDD (50 cm<sup>2</sup>) and stainless steel (50 cm<sup>2</sup>) were used as anode and cathode respectively. PTFE spacer with various thicknesses of 50, 100, 250 and 500  $\mu$ m for micrometric distances as well as 1000 and 3000  $\mu$ m for millimetric distances were implemented. Unless mentioned otherwise, electrolysis was carried out in galvanostatic mode under the polarization of 4 mA cm<sup>-2</sup> (200 mA).

Concentration of Ca and Mg were monitored using ICP-OES. TIC was measured using a TOC-meter V<sub>CSH</sub> TC/TN. LSV, chronoamperometry and chronopotentiometry were performed by intermediary of a potentiostat connected to the electrochemical cell. The electrode potentials were reported as function of Ag-AgCl RE. LSV was carried out sweeping the range of cathodic potentials from 0 to -1.6 V/Ag-AgCl using a scan rate of 10 mV s<sup>-1</sup>. EIS was conducted in galvanostatic mode covering the frequency ranges between 100 kHz to 100 mHz. The perturbation amplitude varied between 5 and 10 mA RMS according to EIS spectra with 10 ppd. The EIS analysis was programmed at regular interval during galvanostatic electrolysis at applied direct current of 200 mA (i.e., 4 mA cm<sup>-2</sup>). The EIS parameters were extracted by means of EEC model using ZSimpwin® commercial software.

Throughout the series of experiments, the  $j_{app}$  was always higher than all  $j_{lim,0}$  values (**Table VI.1**), whatever the configurations used. The  $j_{lim,0}$  value for each electrochemical cell setup was evaluated in similar way as described in Chapter 4 using Eq. (IV.18). The values of  $k_{\rm m}$  specific for each cell configuration were obtained as a result of mass transfer characterization studies conducted in **Chapter 5**. Therefore, across all configurations from millimetric to submillimetric inspected in this chapter, the electrochemical processes were diffusional mass-transfer controlled. It is worth mentioning that while varying  $d_{\text{elec}}$ , the liquid flow rate was also adjusted to have identical electrolyte residence time across different volumes of electrochemical reactors for relevant comparison. The flow rate of 10, 20, 50, 100, 200 and 600 mL min<sup>-1</sup> were used for 50, 100, 250, 500, 1000 and 3000  $\mu$ m  $d_{\text{elec}}$  respectively. A residence time of 0.025 min was maintained across all configurations. Since the electrolyte flow rate was adapted accordingly to the volume of reactor, cross-sectional electrolyte velocity was also kept constant, i.e. in this work 4 m min<sup>-1</sup>, from one configuration to another. Knowing the fact that the  $d_{\rm H}$  (Eq. (V.15)) was modified from one  $d_{\rm elec}$  to another, Reynolds number (Re) (Eq. (V.12)) also increased with the  $d_{\text{elec}}$  as summarized in Table VI.1. Re values could be calculated by considering the physicochemical properties of the effluent provided in the caption of Table V.1. Re values ranged from 7 (50 μm) to 377 (3000 μm), meaning that the hydrodynamic flow of electrolyte was maintained under laminar regime in all investigated cases [42-45]. This could ensure reliable comparisons of the results obtained under different  $d_{\text{elec}}$  setups.

**Table VI.1**. The values of  $j_{\text{lim},0}/j_{\text{app}}$  ratio,  $k_{\text{m}}$ ,  $d_{\text{H}}$  and Re number associated to the investigated  $d_{\text{elec}}$ .  $j_{\text{app}}=4 \text{ mA cm}^{-2}$ , cathode: stainless steel and anode: BDD.  $C_{\text{O}_2,25^{\circ}\text{C}}=2.81\times10^{-4} \text{ mmol L}^{-1}$  and  $v_{\text{L},25^{\circ}\text{C}}=10^{-6} \text{ m}^2 \text{ s}^{-1}$ .

$d_{ m elec}$ / $\mu { m m}$	$k_{\rm m}$ / ${ m m~s}^{-1}$	$j_{\text{lim,0}}/j_{\text{app}} (\ll 1)$	<i>d</i> <sub>H</sub> / mm	Re
50	$6.44 \times 10^{-5}$	0.1746	0.10	7
100	$3.94 \times 10^{-5}$	0.1068	0.20	16
250	$2.75 \times 10^{-5}$	0.0746	0.50	33
500	$2.33 \times 10^{-5}$	0.0632	0.99	66
1000	$1.82 \times 10^{-5}$	0.0492	1.96	130
3000	$1.61 \times 10^{-5}$	0.0436	5.66	377

#### VI.3. Modeling

#### VI.3.1. Kinetics of Mg(OH)<sub>2</sub> and CaCO<sub>3</sub> electro-precipitation

A kinetic model has been proposed to predict the electro-precipitation by considering the influence of  $d_{\text{elec}}$ . The methodology adopted for a single  $d_{\text{elec}}$  (500 µm) in our previous work has been followed and adapted to take into account the electrodes gap [37]. Local alkalization on cathode surface

occurred due to faradaic reactions of reduction of dissolved O<sub>2</sub> (**Eq.** (**II.9**)) and H<sub>2</sub>O (**Eq.** (**II.10**)), while the OH<sup>-</sup> production can be estimated by Faraday's law (**Eq.** (**VI.1**)):

$$O_2 + 2H_2O + 4e^- \rightarrow 4OH^-$$
 (II.9)

$$2H_2O + 2e^- \rightarrow H_2 + 2OH^-$$
 (II.10)

$$r_{\text{OH}^{-}} = +\frac{d[OH^{-}]}{dt} = \frac{v_{\text{O}_{2}}j_{\text{app}}}{n_{\text{O}_{2}}F}\eta_{\text{O}_{2}} + \left(\nu_{\text{H}_{2}\text{O}}\frac{\eta_{\text{H}_{2}\text{O}}}{n_{\text{H}_{2}\text{O}}} - \nu_{\text{H}_{2}}\frac{\eta_{\text{H}_{2}}}{n_{\text{H}_{2}}}\right)\frac{j_{\text{app}}}{F} - 2k_{\text{Mg(OH)}_{2}}[Mg^{2+}][OH^{-}]^{2}$$
(VI.1)

where  $r_{\rm OH^-}$  is the heterogeneous production rate of OH<sup>-</sup> (mol m<sup>-2</sup> min<sup>-1</sup>),  $k_{\rm Mg(OH)_2}$  is the heterogenous third order rate constant associated to the reaction of electro-precipitation of Mg(OH)<sub>2</sub> (0.33 m<sup>7</sup> mol<sup>-2</sup> min<sup>-1</sup> [37]) on cathode surface,  $n_{\rm O_2}$  and  $n_{\rm H_2O}$  are the numbers of electrons participating in Eq. (II.9) and Eq. (II.10) respectively whilst  $v_{\rm O_2}$  and  $v_{\rm H_2O}$  are the stoichiometric coefficient of OH<sup>-</sup> production from Eq. (II.9) and Eq. (II.10) respectively.  $n_{\rm H_2}$  and  $v_{\rm H_2}$  are respectively the number of electrons reacted and stoichiometry to produce H<sub>2</sub> with respect to the stoichiometry of OH<sup>-</sup> production from H<sub>2</sub>O reduction in Eq. (II.10). [OH<sup>-</sup>] and [Mg<sup>2+</sup>] are the molar concentrations of OH<sup>-</sup> and Mg<sup>2+</sup> ions respectively (mol m<sup>-3</sup>),  $j_{\rm app}$  is the applied current density (A m<sup>-2</sup>), F is the Faraday's constant,  $\eta_{\rm O_2}$  and  $\eta_{\rm H_2O}$  are the current efficiencies attributed to faradaic reactions (Eq. (II.9) and Eq. (II.10)). Under the polarization of 4 mA cm<sup>-2</sup>, their values are taken as 0.02% and 99.98% respectively in accordance to Chapter 4.  $\eta_{\rm H_2}$  is the loss of faradaic current efficiency related to concurrent H<sub>2</sub> evolution in Eq. (II.10) on cathode surface. This contribution was introduced through a term denoted as OH<sup>-</sup> hindrance factor ( $\alpha$ ) due to gas bubbles and is defined in Eq. (VI.2):

$$\alpha = \nu_{\rm H_2O} \frac{\eta_{\rm H_2O}}{n_{\rm H_2O}} - \nu_{\rm H_2} \frac{\eta_{\rm H_2}}{n_{\rm H_2}} \tag{VI.2}$$

Equation (VI.1) can then be simplified into Eq. (VI.3):

$$r_{\rm OH^-} = +\frac{d[OH^-]}{dt} = \frac{v_{\rm O_2} j_{\rm app}}{n_{\rm O_2} F} \eta_{\rm O_2} + \alpha \frac{j_{\rm app}}{F} - 2k_{\rm Mg(OH)_2} [Mg^{2+}] [OH^-]^2$$
 (VI.3)

The precipitation rate of Mg(OH)<sub>2</sub> could be estimated using **Eq.** (**IV.4**). The electro-precipitation of CaCO<sub>3</sub> was modeled following the theoretical concentration of interfacial carbonates  $[CO_3^{2-}]_{int}$  (in mol m<sup>-3</sup>) as previously defined as **Eq.** (**IV.12**). The rate of precipitation of CaCO<sub>3</sub> could then be modeled by its kinetic law expressed in **Eq.** (**IV.13**) where the heterogeneous second order rate constant of heterogeneous electro-precipitation  $k_{CaCO_3}$  was taken as  $3.6 \times 10^{-3}$  m<sup>4</sup> mol<sup>-1</sup> min<sup>-1</sup> [37] (**Chapter 4**).

## VI.3.2. Relationship between double layer capacitance, double layer thickness and interelectrode distance

The double layer capacitance ( $C_{\rm DL}$ ) can be expressed in terms of the thickness of electronic double layer ( $d_{\rm DL}$ ) by Eq. (VI.4) [46, 47]:

$$C_{\rm DL} = \frac{\varepsilon_{\rm S} \, \varepsilon_{\rm 0}}{d_{\rm DI}} \tag{VI.4}$$

where  $C_{DL}$  is expressed in F,  $\varepsilon_S$  the dielectric constant of solvent in bulk (here water, the value was taken at 80 [48-50]),  $\varepsilon_0$  the vacuum permittivity (8.854 × 10<sup>-12</sup> F.m [49, 51]) and  $d_{DL}$  is expressed in m.

It is proposed to relate  $d_{\text{elec}}$  with  $d_{\text{DL}}$  using the following expression of Eq. (VI.5):

$$d_{\text{elec}} = \omega \left( d_{\text{DLA}} + d_{\text{DLC}} \right) \tag{VI.5}$$

where  $\omega$  is a constant ( $\omega \gg 1$ ),  $d_{\text{DL,A}}$  and  $d_{\text{DL,C}}$  are the double layer thicknesses on anode and cathode, respectively. Supposing the double layer thickness is identical on both anode and cathode ( $d_{\text{DL,A}} = d_{\text{DL,C}}$ ), Eq. (VI.5) is simplified into Eq. (VI.6) and then into Eq. (VI.7).

$$d_{\text{elec}} = 2 \,\omega \,d_{\text{DL}} \tag{VI.6}$$

$$d_{\rm DL} = d_{\rm elec}/(2\omega) \tag{VI.7}$$

Inserting the expression of  $d_{DL}$  from **Eq.** (VI.7) into **Eq.** (VI.4) gives out a novel equation (**Eq.** (VI.8)):

$$C_{\rm DL} = \frac{\varepsilon_{\rm S} \, \varepsilon_0 \, 2 \, \omega}{d_{\rm elec}} \tag{VI.8}$$

#### VI.3.3. Modeling and fitting evaluation

The modeling of the evolution of Mg<sup>2+</sup> and Ca<sup>2+</sup> concentrations were done using Aquasim© [52]. The validity of experimental impedance spectra was controlled using linear KKT method performed using Zsimpwin® and KKT program developed by University of Twente [53, 54]. The fitting between experimental data and the kinetic models were evaluated using RMSE (Eq. (III.3)) as well as ME (Eq. (III.4)).

#### VI.4. Results and discussion

## VI.4.1. Kinetics and modeling of mineral electro-precipitation at various interelectrode distances

The evolution of concentrations of  $Mg^{2+}$ ,  $Ca^{2+}$  and TIC at different  $d_{elec}$  are illustrated in Fig. VI.1. From Fig. VI.1(a),  $Mg(OH)_2$  was noticed to electro-precipitate the most at 100, 250 and 500  $\mu$ m. As the  $d_{elec}$  increased, less  $Mg(OH)_2$  precipitation occurred. It was observed that  $Mg^{2+}$  in electrolyte precipitated by 85.7  $\pm$  1.3, 91.0  $\pm$  0.1, 91.2  $\pm$  0.9, 87.7  $\pm$  0.5 and 81.0  $\pm$  0.1% after 5-hour electrolysis for  $d_{elec}$  of 100, 250, 500, 1000 and 3000  $\mu$ m respectively. Figures VI.1(b) and VI.1(c) plot the evolution of  $Ca^{2+}$  and TIC concentrations respectively, across multiple  $d_{elec}$  investigated. It can be observed that the  $CaCO_3$  electro-precipitated the most at 100  $\mu$ m and lesser precipitation of  $CaCO_3$  was obtained with increasing  $d_{elec}$  (Fig. VI.1(b) and Fig. VI.1(c)). 55.1  $\pm$  1.1 and 94.8  $\pm$  0.8% of  $Ca^{2+}$  and TIC, respectively, were deposited on cathode using 100  $\mu$ m. These percentages of precipitation decreased to 36.2  $\pm$  2.8 and 82.1  $\pm$  0.4% of  $Ca^{2+}$  and TIC, respectively, when the electrochemical cell was used with 3000  $\mu$ m  $d_{elec}$ . Interestingly, it was noticed that there were no mineral electro-precipitation of  $Mg(OH)_2$  and  $CaCO_3$  at the lowest  $d_{elec}$  investigated (50  $\mu$ m) (Fig. VI.1). These results suggested that the mineral electro-precipitation tended to occur easier when shifting progressively from millimetric into microfluidic setup, except at the lowest  $d_{elec}$  investigated (50  $\mu$ m).

The Mg and Ca mass balances recovery have been estimated to ensure that not only there was no loss of electrolyte during electrolysis (i.e. via leakage or dead volume in the pilot), but also to verify that the trend of the occurrence of mineral cathodic depositions was correct. **Figure VI.2** shows the mass balances of Mg and Ca successively recovered from different parts of experimental pilot after 5 h of electrolysis under all investigated  $d_{\rm elec}$ . The sum of the remaining Mg and Ca in solution with their mass recovered from anode, cathode and tubing remained above  $84 \pm 5\%$  with the average recovery of  $91.2 \pm 2.2\%$  for Mg and  $99.7 \pm 4.1\%$  for Ca across all  $d_{\rm elec}$  investigated. The majority of Mg and Ca precipitated from electrolyte was found on cathode, while little mass was recovered from tubing and anode surface. This observation can confirm the cathodic precipitation of Mg(OH)<sub>2</sub> and CaCO<sub>3</sub> induced by the phenomenon of local alkalization on cathode surface [55-57] and validate the results presented in **Fig. VI.1**.

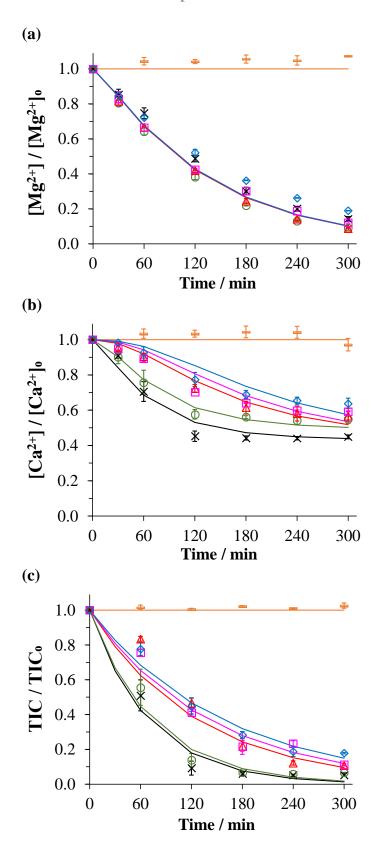
As can be observed in **Fig. VI.1**, the model curves of Mg<sup>2+</sup>, Ca<sup>2+</sup> and TIC fitted well with their experimental counterparts at different  $d_{\text{elec}}$  under study, which confirmed the need to consider gas bubbles evolution through  $\alpha$  parameter as discussed in **Section VI.4.3**. The calculated RMSE values

#### Chapter VI

for Mg<sup>2+</sup>, Ca<sup>2+</sup> and TIC curves fitting across all  $d_{\rm elec}$  are tabulated in **Table VI.2**. RMSE values for Mg<sup>2+</sup> curves fitting was as low as 0.0028 under 500  $\mu$ m configuration, whilst the highest error (0.0149) was found under the 3000  $\mu$ m setup (**Table VI.2**). The lowest and highest RMSE values in Ca<sup>2+</sup> curves fitting were 0.0996 and 0.1983, corresponding to  $d_{\rm elec}$  of 250  $\mu$ m and 1000  $\mu$ m respectively. The kinetics of TIC evolution in the bulk under all  $d_{\rm elec}$  were pseudo first-order rates, as already previously noticed [37]. The lowest RMSE (0.0818) was found for 50  $\mu$ m, whereas the highest RMSE (0.4658) corresponded to the configuration of 500  $\mu$ m.

**Table VI.2**. Model fitting evaluation by RMSE between modeled and experimental curves of  $Mg^{2+}$ ,  $Ca^{2+}$  and TIC concentrations in all investigated  $d_{elec}$  configurations.  $j_{app}$ : 4 mA cm<sup>-2</sup>, cathode: stainless steel and anode: BDD.

d. /		RMSE	
$d_{ m elec}$ / ${ m \mu m}$ -	$\mathrm{Mg}^{2+}$	Ca <sup>2+</sup>	TIC
50	0.0095	0.1041	0.0818
100	0.0086	0.1352	0.2677
250	0.0068	0.0996	0.2810
500	0.0028	0.1026	0.4658
1000	0.0038	0.1983	0.2716
3000	0.0149	0.1704	0.2231



**Fig. VI.1**. Evolution of experimental and modeled concentrations of (a)  $Mg^{2+}$ , (b)  $Ca^{2+}$  and (c) TIC during the electrolysis at 4 mA cm<sup>-2</sup> for different  $d_{elec}$  (—: 50 μm,  $\times$ : 100 μm,  $\bigcirc$ : 250 μm,  $\triangle$ : 500 μm,  $\square$ : 1000 μm and  $\diamondsuit$ : 3000 μm). Cathode: stainless steel and anode: BDD. Solution containing 150 mg  $L^{-1}$  of  $Ca^{2+}$ , 5 mg  $L^{-1}$  of  $Mg^{2+}$  and 60 mg  $L^{-1}$  of TIC.

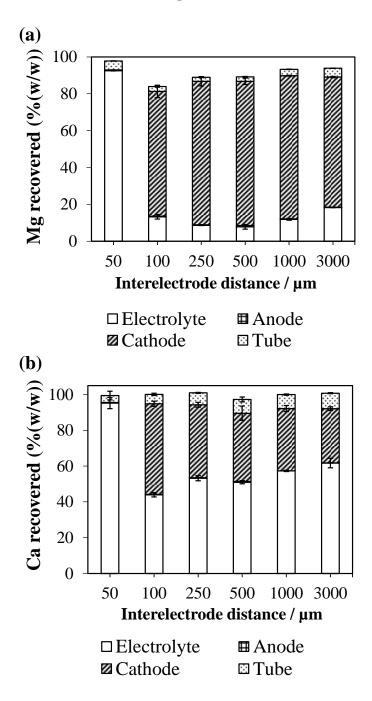


Fig. VI.2. Mass balance of (a) Mg and (b) Ca recovered from different parts of experimental pilot ( $\square$ : electrolyte,  $\square$ : cathode,  $\boxplus$ : anode and  $\square$ : tube) after 5 h of electrolysis under different configurations of  $d_{\text{elec}}$ .  $j_{\text{app}}$ : 4 mA cm<sup>-2</sup>, cathode: stainless steel, anode: BDD and solution: 150 mg L<sup>-1</sup> of Ca<sup>2+</sup>, 5 mg L<sup>-1</sup> of Mg<sup>2+</sup> and 60 mg L<sup>-1</sup> of TIC.

# VI.4.2. Impact of cathode potential on mineral electro-precipitations at different interelectrode distances

The trends observed in Section VI.4.1 can be elucidated by the values of the cathode potential  $(E_{\rm C})$ measured using the potentiostat, when 4 mA cm<sup>-2</sup> current density was applied to the different electrochemical cell configurations. E<sub>C</sub> values along with the average percentages of mass of Mg, Ca and TIC electro-precipitated after 5 h of electrolysis at 4 mA cm<sup>-2</sup> for each  $d_{\rm elec}$  are illustrated in Fig. VI.3(a). It can be noticed that the  $E_{\rm C}$  values dropped significantly from 0.202  $\pm$  0.024 V/Ag-AgCl  $(50 \,\mu\text{m})$  to  $-1.453 \pm 0.021 \,\text{V/Ag-AgCl}$  (100  $\,\mu\text{m}$ ). The diminishing trend continued further down to -1.915 ± 0.030 V/Ag-AgCl when 3000 μm was used. LSV curves corresponding to all investigated  $d_{\text{elec}}$  are presented in Fig. VI.4. Across all interelectrode gaps, the plateau of limiting current of reduction of dissolved O<sub>2</sub> into OH<sup>-</sup> ions (Eq. (II.9)) were noticeable in the potential range of -0.7 to -1.10 V/Ag-AgCl (Fig. VI.4). Comparable range of values have been reported in literature [16, 23, 27-29, 38, 40]. Within these ranges of applied potential, OH<sup>-</sup> production via 4-electron of ORR was the most selective with minimal concomitant gas evolution on electrodes surface. The  $E_{\rm C}$  measured with 100  $\mu$ m (-1.453  $\pm$  0.021 V/Ag-AgCl) was the closest to the potential of ORR into OH<sup>-</sup> (-0.7 to -1.10 V/Ag-AgCl). As a consequence, amongst the various  $d_{\text{elec}}$  under study, the electro-precipitation of Mg(OH)<sub>2</sub> (Eq. (II.11)) and CaCO<sub>3</sub> (Eq. (II.13)) took place with minimal evolving H<sub>2</sub> perturbation when the electrochemical cell was used with  $d_{\rm elec}$  of 100  $\mu$ m. As the  $d_{\rm elec}$  increased,  $E_{\rm C}$  progressively moved farther away into the region of potential of H<sub>2</sub>O reduction which started at about -1.10 V/Ag-AgCl according to Fig. VI.4. Within this region, the OH<sup>-</sup> production still occurred but concomitantly with HER [14, 38-40, 58-60]. As a result, lesser production of mineral electro-precipitation was observed with the increasing interelectrode gap. Furthermore, using the LSV curves plotted in Fig. 4, Tafel slopes [61, 62] corresponding to the region of measured  $E_{\rm C}$  for all  $d_{\rm elec}$  configurations were determined and they are illustrated in Fig. VI.5. Tafel slope corresponding to  $d_{\rm elec}$  of 3000  $\mu m$ depicted the highest slope (-1.57 V dec<sup>-1</sup>) amongst all the investigated  $d_{\rm elec}$ , which accentuated the fact that the highest overpotential was obtained while operating with 3000 µm in comparison with the micrometric distances (-0.38 V dec<sup>-1</sup> using 100 µm).

The higher concurrent  $H_2$  evolution activity when  $d_{\text{elec}}$  increased can be further justified by the ratio of  $j_{\text{lim},0}$  over the adopted  $j_{\text{app}}$ . The ratios are plotted in **Fig. VI.3(b)**. When the  $j_{\text{app}}$  value is inferior or equal to  $j_{\text{lim},0}$  (i.e.,  $j_{\text{lim},0}/j_{\text{app}} \geq 1$ ), it represents a condition in which  $OH^-$  is produced from the 4-electron ORR of **Eq. (II.9)** without any parasitic HER (**Eq. (II.10))** [63, 64]. When  $j_{\text{app}}$  was greater than  $j_{\text{lim},0}$  ( $j_{\text{lim},0}/j_{\text{app}} < 1$ ), it means that  $OH^-$  production was accompanied by concomitant HER due to mass transfer limitation of dissolved  $O_2$ . From **Fig. VI.3(b)**, it can be observed that the  $j_{\text{lim},0}$  values

decreased when  $d_{\rm elec}$  increased. Hence, the  $j_{\rm lim,0}/j_{\rm app}$  was further reduced with increasing  $d_{\rm elec}$  and remained below 1 whatever the distance (**Table VI.1**). Consequently, parasitic HER was more pronounced with increasing  $d_{\rm elec}$ . Contrastingly, under the 50  $\mu$ m  $d_{\rm elec}$  setup, the  $E_{\rm C}$  measured was too low for the 4-electron ORR yet to occur. This enlightened the reason why no precipitation was spotted at this minimum  $d_{\rm elec}$  investigated.

To further confirm the role of  $E_{\rm C}$  in electro-precipitation phenomena at different  $d_{\rm elec}$ , experiments were performed in potentiostatic mode at the same  $E_{\rm C}$  (-1.6 V/Ag-AgCl) - the applied  $E_{\rm C}$  measured on cathode when 4 mA cm<sup>-2</sup> was applied under 500 µm setup - for 500 and 3000 µm interelectrode gap configurations. The electrolyte contained 150 mg L<sup>-1</sup> of Ca<sup>2+</sup>, 5 mg L<sup>-1</sup> of Mg<sup>2+</sup> and 60 mg L<sup>-1</sup> of TIC, identical to that used in Fig. VI.1 and Fig. VI.3. The resulting evolution of Mg<sup>2+</sup>, Ca<sup>2+</sup> and TIC concentrations under the potentiostatic electrolysis are given in Fig. VI.6. It was observed that after 5 h of electrolysis,  $85.2 \pm 2.0\%$  of Mg<sup>2+</sup> precipitated on cathode under submillimetric setup whereas only  $59.2 \pm 0.9\%$  of Mg<sup>2+</sup> from electrolyte was deposited under millimetric setup (Fig. VI.6(a)). At the end of the electrolysis, it was noticed a similar amount of  $Ca^{2+}$  deposited using 500 µm (50.6  $\pm$ 0.1%) in comparison to 3000  $\mu$ m (49.4  $\pm$  0.6%) with however slightly faster kinetics of Ca<sup>2+</sup> precipitation under microfluidic configuration up till 4 h of electrolysis (Fig. VI.6(b)). The TIC evolution in Fig. VI.6(c) suggested that faster consumption of carbonates species occurred under microfluidic setup leading to 93.5  $\pm$  0.2% of CO<sub>3</sub><sup>2-</sup> precipitated with Ca<sup>2+</sup> forming CaCO<sub>3</sub> at 500  $\mu$ m against 89.0  $\pm$  0.6% of  $CO_3^{2-}$  precipitation occurring at 3000 µm. The difference of electroprecipitation observed at both distances and at the same applied  $E_{\rm C}$  indicated that  $E_{\rm C}$  was not the sole influencing factor. The influence of the electrochemical reactor geometry could be also involved. The dissimilarity observed in Fig. VI.6 could then be explained by a difference that could have been brought while transitioning into the microfluidic configuration. Using micrometric  $d_{\rm elec}$ , the rate of OH<sup>-</sup> production in Eq. VI.(3) was intensified by external provision of dissolved O<sub>2</sub> coming from the oxidation of solvent on anode (Eq. (II.3)), producing more OH<sup>-</sup> in comparison to the millimetric  $d_{\text{elec}}$ .

$$2H_2O \rightarrow O_2 + 4H^+ + 4e^-$$
 (II.3)

Consequently, interfacial OH<sup>-</sup> concentration was higher under microfluidic than millimetric setup. Subsequently, thermodynamics of **Eq.** (II.11) and **Eq.** (II.12) were shifted to the right to favor the precipitation reaction of Mg(OH)<sub>2</sub> and the production of CO<sub>3</sub><sup>2-</sup> respectively. Therefore, quicker kinetics and more electro-precipitations of Mg(OH)<sub>2</sub> and CaCO<sub>3</sub> perceived in **Fig. VI.6** are justified. This anodic provision of dissolved O<sub>2</sub> could occur using 3000 µm as well, but it did at much lesser extent. One could question that the production of protons in **Eq.** (II.3) could compensate in the

meantime the interfacial OH<sup>-</sup> at cathode vicinity under microfluidic configuration to produce H<sub>2</sub>O (**Eq. (VI.9**)), which could however be reduced back to OH<sup>-</sup> following **Eq. (II.10**).

$$H^+ + OH^- \rightarrow H_2O \tag{VI.9}$$

Furthermore, the evolution of bulk pH during the electrolysis using different cell configurations is depicted in **Fig. VI.7**. From a practical point of view, it was noticed that within the case where no mineral precipitation was observed, i.e., at 50 µm (**Fig. VI.7**), the bulk pH tended to increase. This behavior was also observed in the previous work when the operating conditions allowed to avoid electro-precipitation at 500 µm interelectrode gap [37]. This shows that the intensification of OH-production due to the anodic source of dissolved O<sub>2</sub> was more predominant than the compensation of protons with OH-, at both interface and in the bulk.

Again, in term of practicability, electrolysis in galvanostatic mode has more interest for the application in the real field. The incoming sections are devoted to further enlighten the mechanism of mineral scaling evolution at different  $d_{\rm elec}$  by exploiting the concomitant gas evolution and characterizing the electrode/electrolyte interface through impedance studies.

(a) Interelectrode distance / µm 100 250 500 3000 50 1000 0.3 Electrolyte precipitated / % 100  $E_{\rm C} / N/Ag - 0.5$ -0.9
-0.9
-1.3 -0.180 60 40 20 -1.7 -2.1  $\square$ Mg<sup>2+</sup> (%)  $\boxtimes$ Ca<sup>2+</sup> (%)  $\blacksquare$ TIC (%)  $\times$ E<sub>C</sub> (V/Ag-AgCl)

## $(b) \longrightarrow \mathbf{E}_{\mathbf{C}}(\%) \longrightarrow \mathbf{E}_{\mathbf{C}}(\mathsf{V}/\mathsf{Ag-AgCI})$

## Interelectrode distance / µm

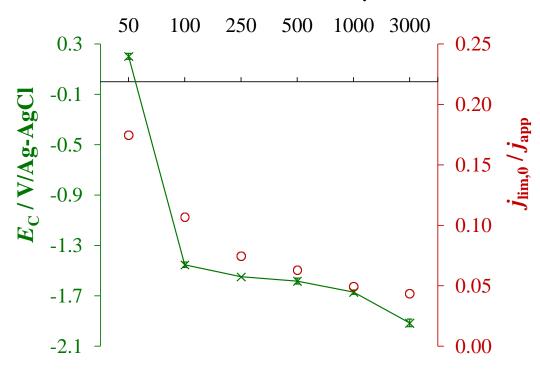
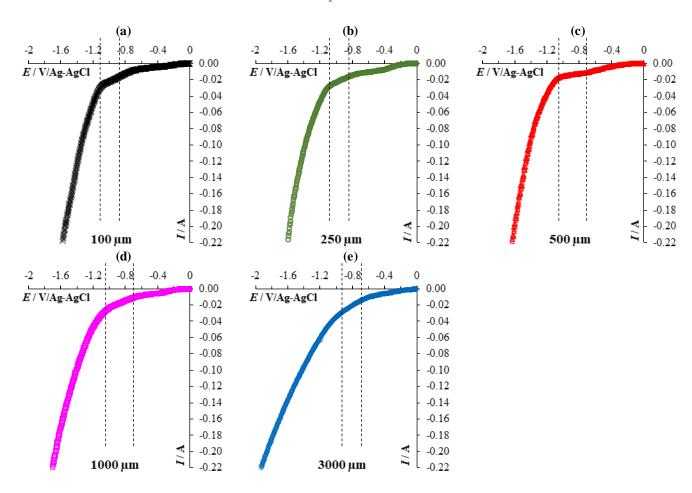
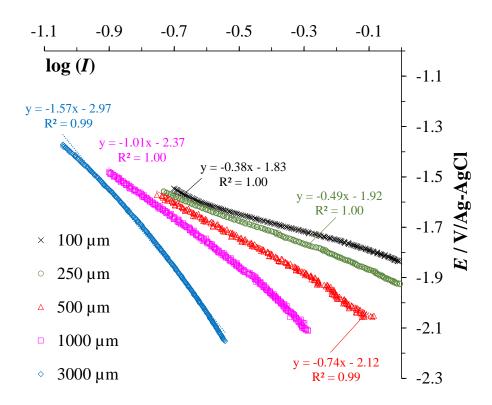


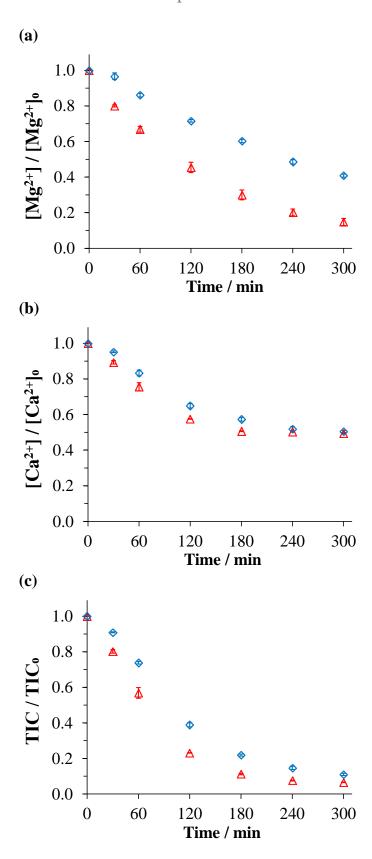
Fig. VI.3. Influence of  $d_{\text{elec}}$  on the evolution of (a)  $E_{\text{C}}$  ( $\times$ ) and average percentage of Mg ( $\square$ ), Ca ( $\boxtimes$ ) and TIC ( $\equiv$ ) electro-precipitated, and (b)  $j_{\text{lim},0}/j_{\text{app}}$  ( $\bigcirc$ ). Cathode: stainless steel, anode: BDD,  $j_{\text{app}}$ : 4 mA cm<sup>-2</sup>. Solution contained 150 mg L<sup>-1</sup> of Ca<sup>2+</sup>, 5 mg L<sup>-1</sup> of Mg<sup>2+</sup> and 60 mg-C L<sup>-1</sup> of TIC.



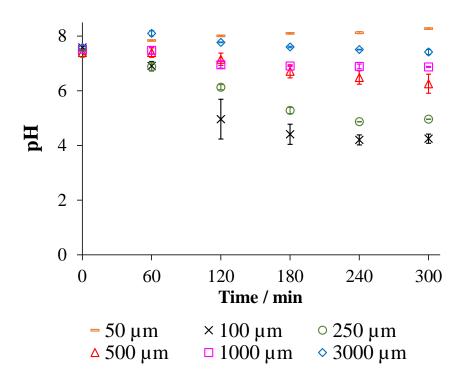
**Fig. VI.4**. LSV of electrochemical reactors with (a) 100 μm, (b) 250 μm, (c) 500 μm, (d) 1000 μm and (e) 3000 μm  $d_{\rm elec}$  in solution containing 150 mg L<sup>-1</sup> Ca<sup>2+</sup>, 5 mg L<sup>-1</sup> Mg<sup>2+</sup> and 60 mg-C L<sup>-1</sup> TIC. Scan rate: 10 mV s<sup>-1</sup>, WE: stainless steel and CE: BDD.



**Fig. VI.5**. Tafel slopes in the range of investigated  $j_{\rm app}$  (4 mA cm<sup>-2</sup>) across different  $d_{\rm elec}$  configurations. Solution contained 150 mg L<sup>-1</sup> of Ca<sup>2+</sup>, 5 mg L<sup>-1</sup> of Mg<sup>2+</sup> and 60 mg-C L<sup>-1</sup> of TIC. WE (cathode): stainless steel and CE (anode): BDD.



**Fig. VI.6**. Evolution of  $Mg^{2+}$ ,  $Ca^{2+}$  and TIC concentrations during the electrolysis of solution containing 150 mg L<sup>-1</sup> of  $Ca^{2+}$ , 5 mg L<sup>-1</sup> of  $Mg^{2+}$  and 60 mg L<sup>-1</sup> of TIC at constant  $E_C$  of -1.6 V/Ag-AgCl under microfluidic (500  $\mu$ m,  $\triangle$ ) and millimetric (3000  $\mu$ m,  $\diamondsuit$ ) configurations. WE: stainless steel and CE: BDD.



**Fig. VI.7**. The evolution of bulk pH over time for different  $d_{\text{elec}}$  under study.  $j_{\text{app}}$ : 4 mA cm<sup>-2</sup>, cathode: stainless steel, anode: BDD and solution: 150 mg L<sup>-1</sup> of Ca<sup>2+</sup>, 5 mg L<sup>-1</sup> of Mg<sup>2+</sup> and 60 mg L<sup>-1</sup> of TIC.

## VI.4.3. The influence of gas evolution on the formation of mineral electroprecipitation

It has been shown that  $\alpha$ , the parmeter accounting for physical gas evolution on cathode surface (**Eqs.** (**VI.2**)-(**VI.3**)), was primordial to yield reliable model fittings to the Mg<sup>2+</sup>, Ca<sup>2+</sup> and TIC experimental curves (**Section VI.4.1**). It was also demonstrated that the more  $d_{\text{elec}}$  increased, the more the ratio of  $j_{\text{lim,0}}/j_{\text{app}}$  decreased due to decreasing  $j_{\text{lim,0}}$  with increasing  $d_{\text{elec}}$  (**Section VI.4.2**). As a consequence, the secondary HER was more vigorous when the  $d_{\text{elec}}$  increased. Subsequently, the mineral electroprecipitation decreased with  $d_{\text{elec}}$ . This feature was also supported by the evolution of  $E_{\text{C}}$  values that decreased progressively farther into the region of water reduction with increasing  $d_{\text{elec}}$  (**Fig. VI.2**). This new interesting insight indicates the impactful influence that the gas evolution has on the formation of mineral electro-precipitation at various  $d_{\text{elec}}$ . Thus, it merits further in-depth investigations as discussed in this sub-section.

The evolution of reduced conductivity of electrolyte ( $\kappa/\kappa_0$ ), which is the ratio of ionic conductivity of electrolyte containing dispersed gas bubbles ( $\kappa$ ) over ionic conductivity of electrolyte free of gas bubbles ( $\kappa_0$ ) [65], is plotted as function of time for different  $d_{\rm elec}$  in Fig. VI.8(a). It has been noticed that  $\kappa/\kappa_0$  decreased the most under the configuration of 100  $\mu$ m, while it increased progressively with  $d_{\rm elec}$ . Therefore,  $\kappa/\kappa_0$  decreased the most under the configuration where the mineral electroprecipitation occurred the most (i.e., 100  $\mu$ m), while when no mineral precipitation was observed (i.e., at 50  $\mu$ m),  $\kappa/\kappa_0$  remained constant over the course of electrolysis. Due to the non-conducting nature of gas bubbles [66, 67], the ionic conductivity of electrolyte is reduced. Using the experimental data of  $\kappa/\kappa_0$ , the equation of Bruggeman written in Eq. (VI.10) allows estimating the gas void ( $\varphi$ ) - the fraction of gas volume over the total volume of liquid and gas - for different  $d_{\rm elec}$  at 4 mA cm<sup>-2</sup>.

$$\varphi = 1 - \left(\frac{\kappa}{\kappa_0}\right)^{2/3} \tag{VI.10}$$

The  $\varphi$  values are plotted against the interelectrode gap in **Fig. VI.8(b)**.  $\varphi$  was noticed to decrease as function of  $d_{\text{elec}}$ , which agreed with the above-mentioned trend of  $\kappa/\kappa_0$  with  $d_{\text{elec}}$  (**Fig. VI.8(a)**). It means, the resistivity due to non-conducting gas bubbles was at the highest under the configuration of 100  $\mu$ m and it decreased progressively with increasing order of  $d_{\text{elec}}$ . The higher values of  $\varphi$  obtained in smaller interelectrode gaps was confirmed by several works in literature [66, 68].

The term relative resistivity  $(\rho/\rho_0)$  (**Eq. (VI.11**)) is also used in literature to depict the evolution of electrolyte resistivity, which is simply the inverse of  $\kappa/\kappa_0$ . The  $\rho/\rho_0$  is plotted in function of  $d_{\text{elec}}$  in **Fig. VI.8(c)**.

$$\frac{\rho}{\rho_0} = \frac{1}{\kappa_{\kappa_0}} \tag{VI.11}$$

 $\rho/\rho_0$  decreased in the increasing order of  $d_{\rm elec}$ . Comparable trend of the evolution of relative resistivity as function of interelectrode gap has also been highlighted in the work of Hine et al. [69, 70] and Vogt et al. [71] but in a larger range of interelectrode gap (5 to 70 mm) and relatively larger applied current densities (75 to 375 mA cm<sup>-2</sup>). To further assess the relation between the gas evolution, the electrolyte resistivity and  $\alpha$  coefficient, the  $\alpha$  values are plotted against  $d_{\rm elec}$  in Fig. VI.8(c). It can be observed that  $\alpha$  decreased with the increasing  $d_{\rm elec}$ . In agreement with the plots in Fig. VI.8,  $\alpha$  followed the trends of  $\varphi$  and  $\rho/\rho_0$  with  $d_{\rm elec}$ . Therefore, it strengthens the significance of introducing the  $\alpha$  factor to account for physical H<sub>2</sub> evolution during the mineral scaling. In the nutshell, the relationship between the gas evolution, gas void and relative resistivity with the formation of mineral scaling on cathode surface at different interelectrode gaps can be schematized in Fig. VI.9. The least electro-deposition took place at the highest  $d_{\rm elec}$  under study (3000  $\mu$ m) owing to vigorous evolving gas whilst the mineral scaling occurred the most under the 100  $\mu$ m configuration.

The internal ohmic resistance due to ohmic drop ( $\sum RI$ ) in an electrochemical cell not only varies with the variation of interelectrode gap but is also affected by the evolving gas bubbles which act as moving electrical insulators in-between electrodes [67]. Therefore, it appears also interesting to study the evolution  $\sum RI$  that can be evaluated by Eq. (VI.12).

$$\sum RI = \Delta U - |E_{\rm C}| - |E_{\rm A}| \tag{VI.12}$$

where  $\sum RI$  is the ohmic drop (V), while  $|E_C|$  and  $|E_A|$  are the absolute values of cathode and anode potentials (V) respectively.

The plots of  $\sum RI$  and overall  $\Delta U$  as function of  $d_{\rm elec}$  are given in **Fig. VI.10**. It can be clearly observed that, as the  $d_{\rm elec}$  decreased, the ohmic drop and cell voltage decreased linearly down to 100  $\mu$ m. Interestingly, the decrease of  $\sum RI$  occurred abruptly below 100  $\mu$ m, breaking the linearity observed in the range of 100  $\mu$ m to 3000  $\mu$ m configuration. This is the first time that the non-linearity of ohmic resistance is assessed under the microfluidic setup unlike the linear decrease of  $\sum RI$  usually observed using millimetric distances in literature (**Eq. (VI.13**)) [72].

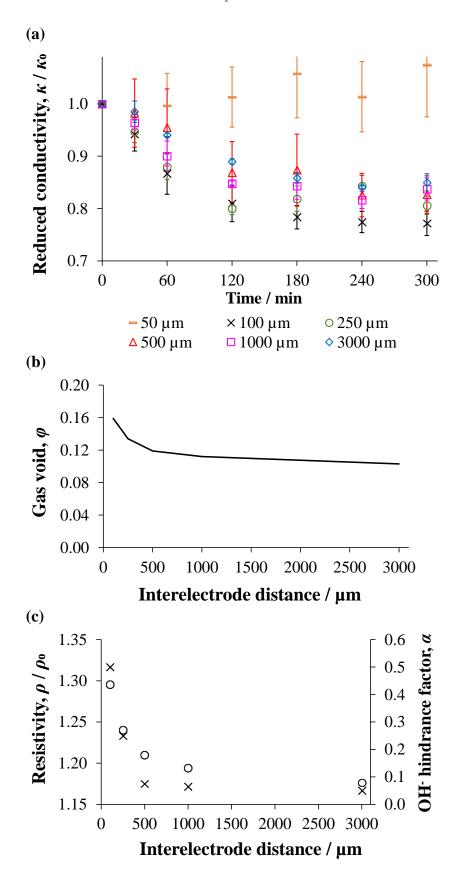
$$\sum RI = \frac{j_{\text{app}} d_{\text{elec}}}{\kappa_0}$$
 (VI.13)

In the presence of gas bubbles, **Eq. (VI.13)** is slightly modified to take into consideration the gas void fraction as shown in **Eq. (VI.14)** [69, 71]:

$$\sum RI = \frac{j_{\text{app}}}{\kappa_0 (1 - \varphi)^{3/2}} d_{\text{elec}}$$
 (VI.14)

The slope value obtained from the linear trend in **Fig. VI.10**, for distances ranging from 100 to 3000  $\mu$ m, equaled 0.2435 V mm<sup>-1</sup> (R<sup>2</sup> = 0.95).

In conclusion, the  $\sum RI$  decreased with decreasing  $d_{\rm elec}$  and higher  $\varphi$  fraction in submillimetric region made the  $\sum RI$  decrease non-linearly. A decrease in electrolytic  $\sum RI$  reduces indeed the energy consumption of an EAOP, nevertheless it does not promise a decrease in mineral scaling formation as practically proven in this work. More mineral precipitation occurred, while the  $\sum RI$  decreased in descending order of  $d_{\rm elec}$  down to 100  $\mu$ m.



**Fig. VI.8**. (a) Evolution of  $\kappa/\kappa_0$  of electrolyte in-between different  $d_{\rm elec}$ , (b) influence of  $d_{\rm elec}$  on the estimated  $\varphi$  fraction and (c) variation of  $\rho/\rho_0$  (O) and  $\alpha$  (×) with  $d_{\rm elec}$ . Cathode: stainless steel, anode: BDD and  $j_{\rm app} = 4$  mA cm<sup>-2</sup>.

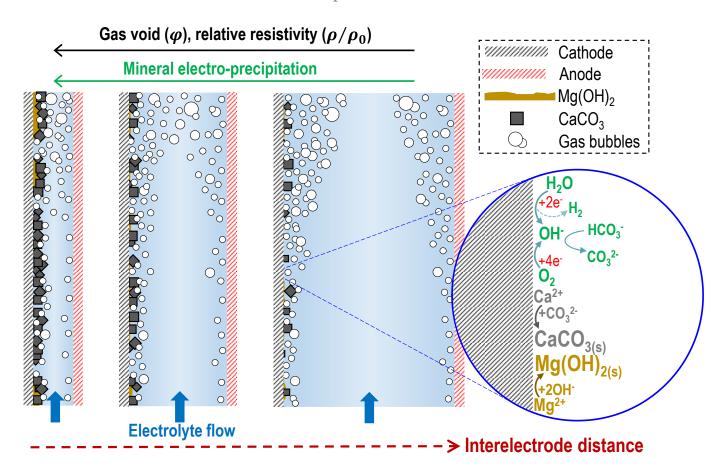


Fig. VI.9. Mechanistic scheme describing the relationship between mineral electro-precipitation with the occurring concurrent gas evolution on cathode surface under different  $d_{\text{elec}}$  configurations.

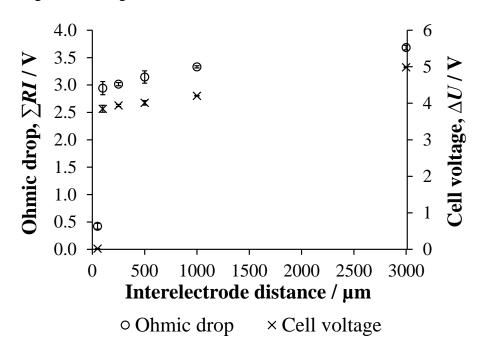


Fig. VI.10. Ohmic drop ( $\sum RI$ ) of electrolyte (O) and cell voltage ( $\Delta U$ ) ( $\times$ ) as a function of  $d_{\text{elec}}$ . Cathode: stainless steel, anode: BDD and  $j_{\text{app}} = 4 \text{ mA cm}^{-2}$ .

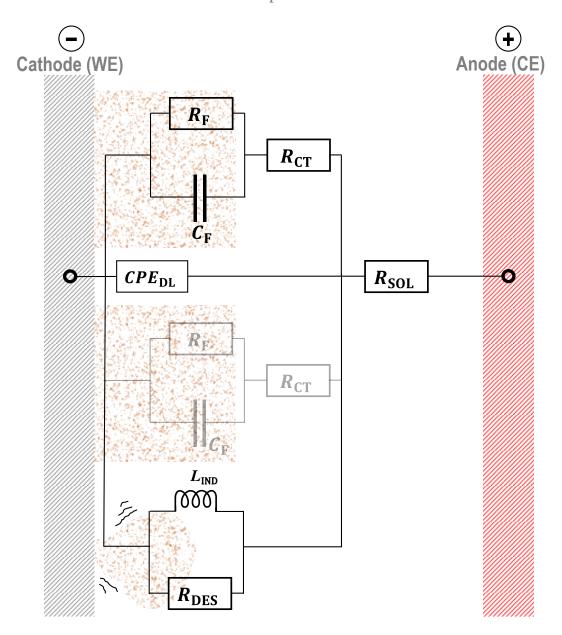
# VI.4.4. Cathode/electrolyte interface impedance study of electro-precipitation at different interelectrode distances

It is presented here-in for the first time, the interfacial parameters acquired by EIS analysis of a scalable electrochemical reactor in the aim at better understanding the cathode/electrolyte interface mechanism under electro-precipitation condition considering wide ranges of  $d_{\rm elec}$ . The EIS extracted parameters under distinct configurations of millimetric into submillimetric setup are compared and discussed. A mechanistic scheme of mineral electro-precipitation on cathode surface is proposed with the integrated EEC to characterize the precipitation using EIS (Fig. VI.11).  $R_F$  and  $C_F$  corresponded respectively to the resistance and capacitance of porous non-conductive precipitation film. Charge transfer resistance (R<sub>CT</sub>) was attributed to the difficulty of electronic transfer for electrochemical reactions to occur on cathode surface as the mineral electro-precipitation grew over time. R<sub>SOL</sub> was attributed to the resistance of electronic transfer in electrolyte between cathode (as working electrode) and reference electrode. Instead of pure capacitor circuit element, constant phase element ( $CPE_{DL}$ ) was used to model the charging of non-faradaic capacitive electronic behavior that occurred on bare cathode surface - the area where there was no electro-precipitation - with ions present in electrolyte. This capacitive response was due to the electronic double-layer formed at the polarized interface. CPE<sub>DL</sub> was more apt to account for not only the non-homogeneity of distribution of metallic elements that composed stainless steel cathode [73-75] but also non-ideality of capacitive response due to uneven coverage of porous electro-precipitation and evolving gas on cathode surface [28, 29, 76, 77]. The value of  $C_{DL}$  is linked with  $CPE_{DL}$  (in F) by Eq. (VI.15) [74, 76-78]:

$$CPE_{DL} = C_{DL}^{n} [R_{SOL}^{-1} + R_{CT}^{-1}]^{(1-n)}$$
 (VI.15)

where n (0 < n < 1) is the constant phase element parameter and  $R_{SOL}$  as well as  $R_{CT}$  are expressed in Ohms.

Another time-constant,  $L_{\text{IND}}$ - $R_{\text{DES}}$  (L in Henry and  $R_{\text{DES}}$  in Ohms) was added to complete the equivalent circuit. This pair of time constant is ascribed to  $R_i$ - $C_i$  couple when  $R_i$  and  $C_i$  take negative values, as the outcome of mathematical fitting of equivalent circuit to experimental data. This is often used when induction loop appears in the experimental data, which was also noticed previously not only in batteries, fuel cells and corrosion study but also in water electrolysis system with gas evolving electrodes [79-83]. Consequently, the inductive loop in our experimental spectra has been attributed to the possible adsorption/desorption of mineral electro-precipitate at the cathode surface.



**Fig. VI.11**. Mechanistic scheme of mineral electro-precipitation on cathode surface including the representation of EEC used in EIS modeling. WE: stainless steel and CE: BDD.

The validity of the impedance spectra obtained in our experiments was checked and validated by the linear KKT method [84-87] and they were plotted against the experimental data at varying  $d_{\text{elec}}$  as shown in Fig. VI.12. Negative impedance values obtained in high frequency regions were not considered as they were the outcome of instrumental artifacts. The chi-squared residues  $\chi^2$  between linear KKT and EEC model fitting curves against experimental data are tabulated in Table VI.3. The experimental curves complied well with the KKT plots considering the highest  $\chi^2$  deviation of 0.0017, whereas the modeled curves fitted the experimental counterparts very well with the highest  $\chi^2$  of 0.0103. The conformity of experimental EIS spectra with KKT standard confirmed the proposed EEC model (Fig. VI.11).

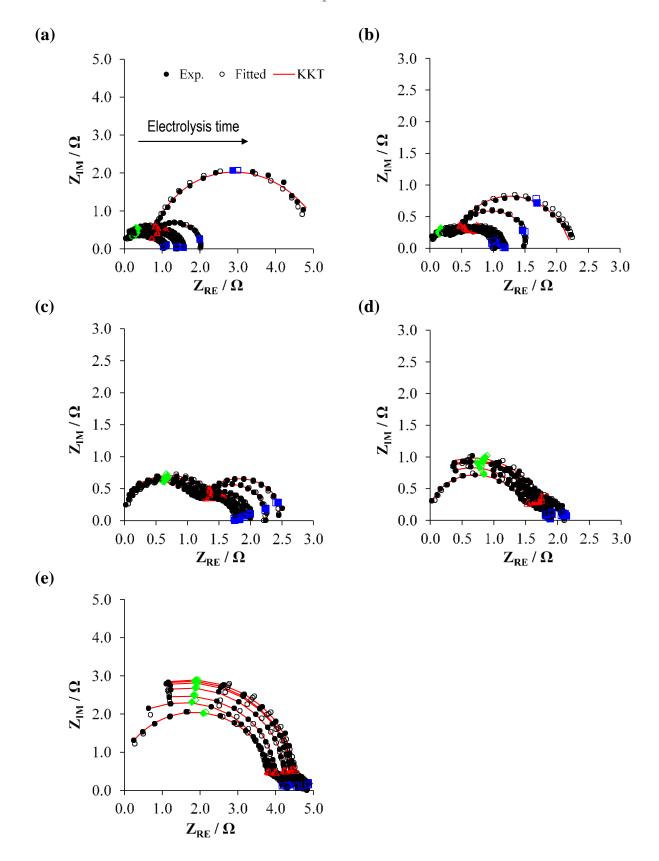


Fig. VI.12. Plots of linear KKT (—), experimental (●) and EEC theoretical (○) impedance spectra for d<sub>elec</sub> of (a) 100 μm, (b) 250 μm, (c) 500 μm, (d) 1000 μm and (e) 3000 μm. (◆, ⋄): high frequency (20 kHz), (△, △): middle frequency (2 kHz) and (□, □): low frequency (0.2 kHz) impedance samples. Electrolysis time: 0, 0.5, 1, 2, 3, 4 and 5 h.

**Table VI.3**. Chi-squared ( $\chi^2$ ) residues between EIS experimental spectra with linear KKT control and EEC theoretical plots for different investigated  $d_{\text{elec}}$ .

Time /	100 µm		250 μm		500 μm		1000 μm		3000 μm	
h	KKT	EEC								
0	0.00043	0.00187	0.00058	0.01028	0.00016	0.00188	0.00048	0.00256	0.00002	0.00082
0.5	0.00011	0.00389	0.00058	0.00537	0.00011	0.00284	0.00018	0.00121	0.00003	0.00238
1	0.00029	0.00362	0.00063	0.00418	0.00005	0.00352	0.00016	0.00431	0.00003	0.00128
2	0.00031	0.00450	0.00071	0.00708	0.00007	0.00716	0.00010	0.00165	0.00003	0.00181
3	0.00031	0.00337	0.00060	0.00441	0.00007	0.00203	0.00026	0.00249	0.00004	0.00229
4	0.00023	0.00285	0.00170	0.00950	0.00004	0.00544	0.00016	0.00163	0.00004	0.00298
5	0.00033	0.00220	0.00130	0.00610	0.00007	0.00494	0.00010	0.00162	0.00004	0.00656

Figure VI.13 depicts the evolution of  $R_{CT}$ ,  $R_F$ ,  $C_{DL}$  and  $C_F$  over time for all the investigated  $d_{elec}$ .  $C_{DL}$ values estimated correspond well to the average values widely reported in literature (4-60 µF) [27, 39, 41, 51, 78, 88, 89]. An interesting feature is that  $R_{CT}$  and  $C_{DL}$  values were seen to be constant over the course of experiment, whatever the  $d_{\text{elec}}$  adopted (Fig. VI.13). The average values of  $R_{\text{CT}}$  were  $0.79\pm0.04~\Omega,~0.55\pm0.05~\Omega,~1.1\pm0.14~\Omega,~1.6\pm0.06~\Omega$  and  $4.4\pm0.18~\Omega$  for 100, 250, 500, 1000 and 3000  $\mu$ m  $d_{\rm elec}$  respectively. The average  $C_{\rm DL}$  values were  $25\pm1.7~\mu$ F,  $21\pm2.4~\mu$ F,  $8\pm1.3~\mu$ F,  $6.2\pm0.2$  $\mu$ F and 2.6±0.1  $\mu$ F for respective distances of 100, 250, 500, 1000 and 3000  $\mu$ m. In literature,  $R_{\rm CT}$ and  $C_{\rm DL}$  have been reported to respectively increase and decrease as function of electrolysis time [17, 21, 22, 27, 41, 78, 79], due to non-conductive nature of CaCO<sub>3</sub> and Mg(OH)<sub>2</sub> deposits which progressively cover the effective surface of electrode. Contrastingly, under this experimental setup, the progressive formation of mineral electro-precipitate did not significantly impede the electrochemical reactions occurring on the electrode/electrolyte interface. This observation could be attributed to the large cathode surface used in these experiments, which was about 25 to 250 times bigger than the classic rotating disk electrodes used by others [37]. The cathode surface was not fully covered by the precipitate and hence offered adequate sites for electronic exchanges at electrode/electrolyte interface. Moreover, owing to concomitant gas evolution on the cathode, the morphology of electro-precipitate is known to be porous which hindered the complete coverage of cathode surface [39, 90]. Lastly, the overall trend of the impedance response attributed to the faradaic resistance of non-conducting porous layer  $(R_{\rm F})$  was found to increase over time according to Fig. VI.13. The increase was more pronounced under the configuration in which more mineral electroprecipitation occurred, i.e., when  $d_{\text{elec}}$  was decreasing. After 5 h of electrolysis, there was an increase of about 1.1, 1.3, 2.1, 2.5 and 3.7 times of R<sub>F</sub> relatively to their initial values before electrolysis for the respective distances of 3000, 1000, 500, 250 and 100 µm. In the meantime, C<sub>F</sub> appeared to be rather constant during the electrolysis considering standard deviations, similarly to the evolution of  $C_{\rm DL}$  with electrolysis time (**Fig. VI.13**).

Fig. VI.14 illustrates the experimental and modeled plots of  $R_{\rm CT}$  and  $C_{\rm DL}$  as a function of  $d_{\rm elec}$  at the beginning and after 1, 3 and 5 h of electrolysis. Interestingly,  $R_{\rm CT}$  increased linearly with increasing  $d_{\text{elec}}$  according to Figs. VI.14(a), VI.14(c), VI.14(e) and VI.14(g). At t=0, the linear curve has a slope of 1.20  $\Omega$  mm<sup>-1</sup> (R<sup>2</sup> = 0.99) and after 5 h electrolysis, the slope was evaluated at 1.43  $\Omega$  mm<sup>-1</sup>  $(R^2 = 0.98)$ . One could expect that the more the electrode is covered by non-conducting film, the more the  $R_{\rm CT}$  resistance for faradaic electron exchange at electrode/electrolyte is enhanced. This reverse trend can be firstly explained by the manifestation of varying the  $d_{\rm elec}$ . It has been elaborated from Fig. VI.6 that the involvement of anodic activity producing dissolved O<sub>2</sub> was more pronounced in microfluidic configuration. As a result, R<sub>CT</sub> measured by EIS at narrow interelectrode gap was lower. Secondly, as depicted in Fig. VI.3, when  $d_{\text{elec}}$  increased, the  $E_{\text{C}}$  measured on cathode surface decreased lower into the water reduction region. More vigorous HER occurring at higher  $d_{\text{elec}}$ impeded more effectively the electronic charge transfer. As a result, R<sub>CT</sub> increased as the interelectrode gap increased. Those are the reasons why largest  $R_{\rm CT}$  was characterized with 3000  $\mu m$ setup in Fig. VI.14 despite the least mineral scaling occurred under this configuration. Moreover, the fact that  $R_{CT}$  decreased when  $E_{C}$  increased (i.e., when  $d_{elec}$  decreased) does corroborate the result obtained by Hoseinieh et al. [29]. Eventually, the trend of  $R_{\rm CT}$  as function of  $d_{\rm elec}$  offers a new interesting insight, since it has never been studied elsewhere in the literature and therefore constitutes a baseline for future investigation.

To further explore the trend between  $R_{\rm CT}$  and the  $d_{\rm elec}$ , the slopes of  $R_{\rm CT}$  curves ( $R_{\rm CT}/d_{\rm elec}$ ) obtained in Figs. VI.14(a), VI.14(c), VI.14(e) and VI.14(g) have been plotted as function of electrolysis time (Fig. VI.15). It was noticed that the  $R_{\rm CT}/d_{\rm elec}$  ratio remained practically constant and it took an average value of  $1.32\pm0.11~\Omega$  mm<sup>-1</sup> during the electrolysis. This finding suggested that the electrochemical setup adopted in this work was not prone to prompt total electrode passivation by mineral scaling, at least after 5 h of electrolysis. This is probably due to the large effective surface area employed in this work and further highlights its interest for the application at larger scale.

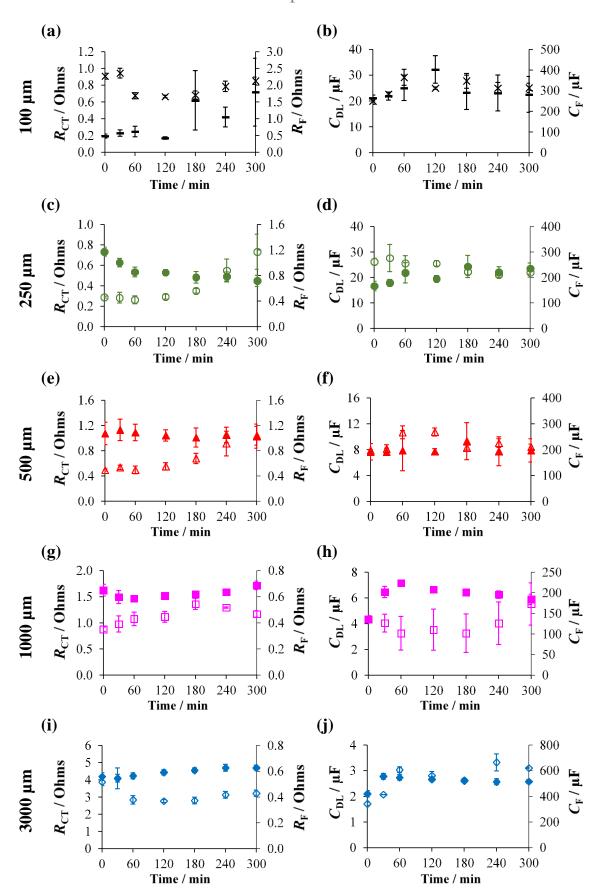
Looking at the redistribution of charges at the electrode/electrolyte interface (**Fig. VI.14(b)**, **VI.14(d)**, **VI.14(f)** and **VI.14(h)**), another new interesting insight was noticed.  $C_{DL}$  values decreased drastically with increasing  $d_{elec}$ . This could partly be attributed to the decrease in the applied  $E_C$  on cathode. As the consequence of more vigorous gas evolution at lower  $E_C$ , the distribution of charges at cathode/electrolyte interface was perturbed and ionic species and their solvated forms could undergo rearrangement at the interface [91-93]. As a result, the capacitance response via EIS measurement was smaller. The apparent decreasing trend of  $C_{DL}$  values with decreasing  $E_C$  was supported by the literature [39, 76, 77, 81, 94]. Furthermore, a new model is proposed to establish the relationship

between  $C_{DL}$  and  $d_{elec}$ .  $C_{DL}$  model curves are plotted as a function of  $d_{elec}$  in **Fig. VI.14(b)**, **VI.14(d)**, **VI.14(f)** and **VI.14(h)** before (t = 0) and after 1, 3 and 5 h of electrolysis using **Eq. (VI.16)**.

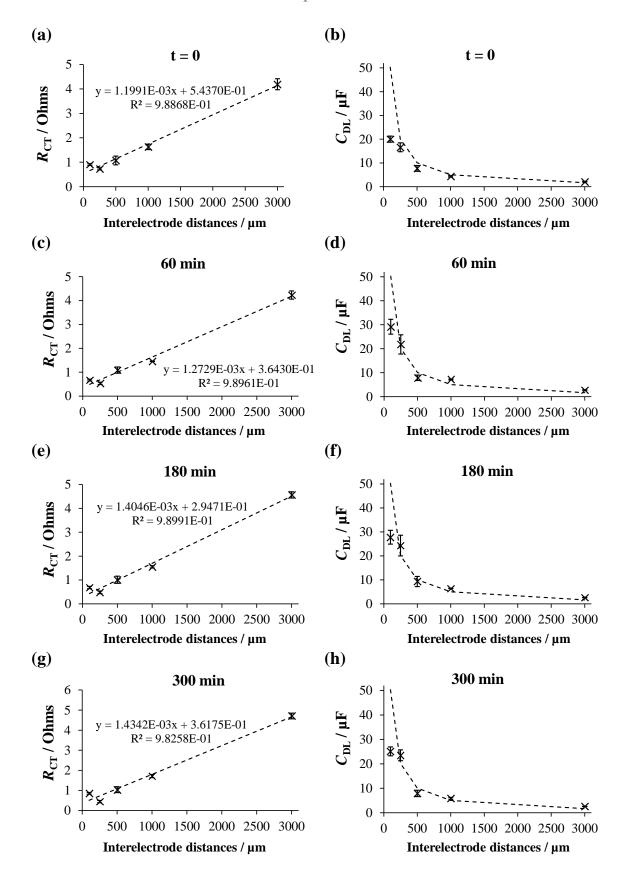
$$C_{\rm DL} = 5.0305 \pm 0.6073 \times \frac{1}{d_{\rm elec}}$$
 (VI.16)

In Eq. (VI.16),  $C_{DL}$  is expressed in  $\mu F$  and  $d_{elec}$  is in mm. It can be observed that  $C_{DL}$  varied linearly with  $1/d_{elec}$  from millimetric distances down to 250  $\mu$ m. Within this range of  $d_{elec}$ , the theoretical  $C_{DL}$  curves matched well the experimental plots. Using ME fitting criterion (Eq. (III.4)), the ME was evaluated as high as 97.93% with the worst fitting scenario calculated to be 95.46%. Below 250  $\mu$ m, the evolution of  $C_{DL}$  was no longer proportional to the inverse of  $d_{elec}$ . More mineral scaling occurring under this condition might be the factor that broke the linearity with  $1/d_{elec}$ . By identification between Eq. (VI.16) and Eq. (VI.8), the slope of the former equals to  $2\varepsilon_S\varepsilon_0\omega$ . Thus, the constant  $\omega$  can be estimated and equaled to  $3.55 \times 10^6$ . Computing Eq. (VI.16) with the approximated  $\omega$  yielded  $d_{DL}$  values of 0.07, 0.14, 0.28 and 0.85 nm when the electrochemical cell was equipped with 250, 500, 1000 and 3000  $\mu$ m  $d_{elec}$  respectively. These values were lower than 3 to 5 nm reported in diluted aqueous solution (electrolyte of 10 mM) at 25°C [48, 95]. This difference might be explained by the concurrent gas evolution taking place in the present case on both cathode and anode surfaces through the entire electrolysis.

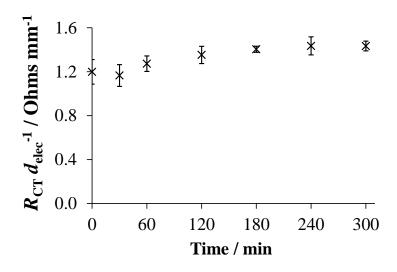
Contrastingly, the evolutions of  $R_F$  and  $C_F$  which are characteristic of insulating deposit films formed on cathode surface in function of  $d_{\text{elec}}$  (**Fig. VI.16**) were more straightforward. Both resistance and capacitance related to the film formation (i.e.,  $R_F$  and  $C_F$ , respectively) decreased in the increasing order of  $d_{\text{elec}}$ . As depicted in **Fig. VI.1**, less mineral deposition took place in the increasing order of interelectrode gap. Consequently, fitted  $R_F$  and  $C_F$  values decreased owing to lesser mineral electroprecipitation formed in ascending order of  $d_{\text{elec}}$ . This further emphasizes the decreasing trends of  $R_F$  and  $C_F$  values with decreasing applied potential which have been reported in literature [27, 39, 76, 78, 81].



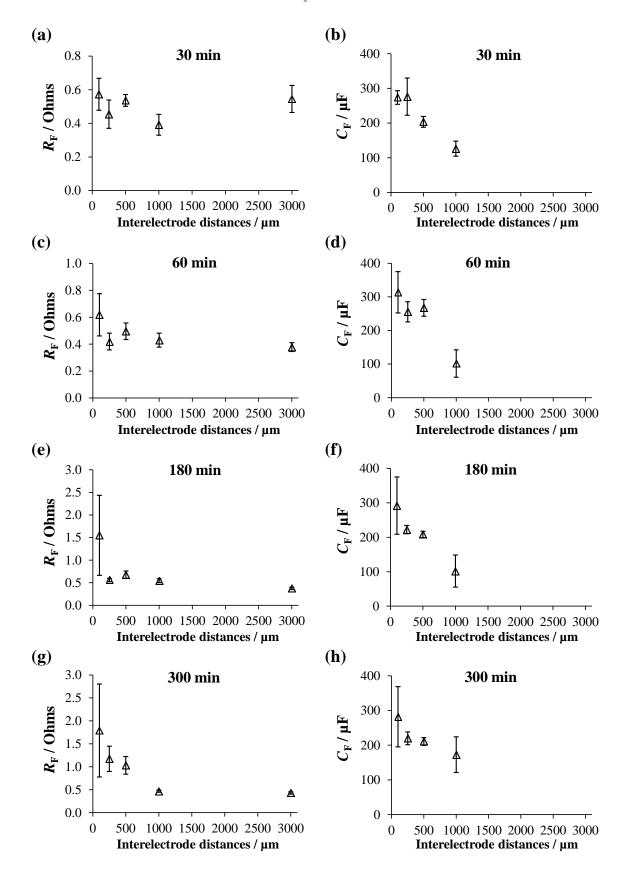
**Fig. VI.13**. Evolutions of  $R_{\text{CT}}$  ( $\times$ ,  $\bullet$ ,  $\blacktriangle$ ,  $\bullet$ ),  $R_{\text{F}}$  (-,  $\bigcirc$ ,  $\triangle$ ,  $\square$ ,  $\diamondsuit$ ) in (a), (c), (e), (g), (i) as well as  $C_{\text{DL}}$  ( $\times$ ,  $\bullet$ ,  $\blacktriangle$ ,  $\bullet$ ) and  $C_{\text{F}}$  (-,  $\bigcirc$ ,  $\triangle$ ,  $\square$ ,  $\diamondsuit$ ) in (b), (d), (f), (h), (j) during 5 h of electrolysis using different  $d_{\text{elec}}$ . Cathode: stainless steel, anode: BDD and  $j_{\text{app}}$ : 4 mA cm<sup>-2</sup>. Solution containing 150 mg L<sup>-1</sup> of Ca<sup>2+</sup>, 5 mg L<sup>-1</sup> of Mg<sup>2+</sup> and 60 mg L<sup>-1</sup> of TIC.



**Fig. VI.14.** Experimental and modeled curves of  $R_{\rm CT}$  in (a), (c), (e) and (g) as well as  $C_{\rm DL}$  in (b), (d), (f) and (h) as a function of  $d_{\rm elec}$  at the initial time and after 1, 3 and 5 h of electrolysis. Cathode: stainless steel, anode: BDD and  $j_{\rm app}$ : 4 mA cm<sup>-2</sup>. Solution containing 150 mg L<sup>-1</sup> of Ca<sup>2+</sup>, 5 mg L<sup>-1</sup> of Mg<sup>2+</sup> and 60 mg L<sup>-1</sup> of TIC.



**Fig. VI.15**.  $R_{\rm CT}/d_{\rm elec}$  slope as a function of electrolysis time.



**Fig. VI.16**. Evolutions of  $R_F$  in (a), (c), (e) and (g) as well as  $C_F$  in (b), (d), (f) and (h) as a function of  $d_{\text{elec}}$  after 0.5, 1, 3 and 5 h of electrolysis. Cathode: stainless steel, anode: BDD,  $j_{\text{app}}$ : 4 mA cm<sup>-2</sup> and solution: Ca<sup>2+</sup> 150 mg L<sup>-1</sup>, Mg<sup>2+</sup> 5 mg L<sup>-1</sup> and TIC 60 mg L<sup>-1</sup>.

#### VI.5. Conclusions

The influence of a wide range of  $d_{\rm elec}$  (50-3000 µm) towards the occurrence of mineral electroprecipitation has been elucidated for the first time. Under similar diffusional mass transfer regime and laminar hydrodynamic regime from one configuration to the others, more deposition of Mg(OH)<sub>2</sub> and CaCO<sub>3</sub> were observed in decreasing order of  $d_{\text{elec}}$  from 3000 to 100 µm, whilst no deposit was present at the lowest  $d_{\text{elec}}$  investigated (50 µm). This new insight was mostly accounted for the applied E<sub>C</sub> when the electrochemical cells were polarized at 4 mA cm<sup>-2</sup>. No OH<sup>-</sup> was yet produced under the 50  $\mu m$  configuration, due to the high  $E_{\rm C}$  value (0.202  $\pm$  0.024 V/Ag-AgCl). From 100  $\mu m$   $d_{\rm elec}$ upwards, the OH- production occurred concomitantly with HER and it was more intense at higher interelectrode gap. This finding was in line with the decreasing measured  $E_{\rm C}$  in ascending order of delec. In this case, the parasitic HER reduced the current efficiency of the formation of OH via dissolved O2 and H2O reduction reactions and also induced detachment of mineral scaling occurring on cathode surface. When identical  $E_{\rm C}$  was applied under millimetric and microfluidic configurations, a difference of electro-precipitation related to the electrochemical reactor design was emphasized. More provision of dissolved O<sub>2</sub> produced from OER on anode contributed towards the formation of mineral scaling on cathode under microfluidic than millimetric configuration. A novel mathematical modeling taking into account the concomitant H<sub>2</sub> evolution to model the kinetics of Mg and Ca electro-precipitation has been proposed. It fitted well with the experimental data (RMSE < 0.0149 for  $Mg^{2+}$  and RMSE < 0.1983 for  $Ca^{2+}$ ). The concurrent gas bubbles formed while varying  $d_{elec}$  showed that both relative resistivity and gas void fraction increased when  $d_{\rm elec}$  decreased from 3000  $\mu m$  to 100  $\mu$ m. Despite the concomitant HER, when  $d_{\text{elec}}$  was reduced, the internal ohmic resistance dropped linearly with decreasing interelectrode gap from millimetric down to 100 µm, but below 100 µm, the linear trend was no longer observed.

Electrochemical processes taking place at the cathode/electrolyte interface were accessed by EIS method. Under the configurations where mineral deposition were observed, it was noticed that the  $R_{\rm CT}$  and  $C_{\rm DL}$  values were constant during the 5 h of electrolysis whatever the  $d_{\rm elec}$  adopted. It was attributed to the large effective surface area of the cathode whereby it was not fully passivated under the applied operating conditions in this work. Contrastingly,  $R_{\rm CT}$  and  $C_{\rm DL}$  values increased and decreased respectively with increasing  $d_{\rm elec}$ . Under microfluidic setups,  $R_{\rm CT}$  values were lower owing firstly to the facility of OH<sup>-</sup> production due to external provision of dissolved O<sub>2</sub> from the oxidation of water on anode, and secondly to the less evolving H<sub>2</sub> at lower  $d_{\rm elec}$ . Using higher  $d_{\rm elec}$ , higher intensity of HER provoked more perturbation of charge distribution at electrode/electrolyte interface, which resulted in lower  $C_{\rm DL}$  values. Novel mathematical relations to assess  $R_{\rm CT}$  and  $C_{\rm DL}$  as a function of  $d_{\rm elec}$  have been proposed to predict their evolution.

#### Chapter VI

Finally, it has been newly demonstrated the importance of  $d_{\text{elec}}$  on the formation of mineral scaling on the cathode surface. Both the effects of  $H_2$  overpotential and electrochemical design (i.e., the contribution of anode under micrometric distances) have been proven influential to hinder or minimize the occurrence of mineral precipitation on cathode surface through the occurrence or not of gas bubbles electrogeneration. Due to primordial role of gas evolution taking an effect on mineral electro-precipitation observed in this work, the overpotential of  $H_2$  evolution offered by other cathode materials could either minimize or maximize the electro-precipitation, which will be the subject of research in incoming chapter.

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# VII. Effect of simulated and real wastewaters on the occurrence of electro-precipitation and organic pollutants degradation

#### VII.1. Introduction

In previous chapters, parametric and modeling studies have been carried out in the electrolyte consisting of only Ca<sup>2+</sup>, Mg<sup>2+</sup> and TIC (except **Chapter 5**) during electro-oxidation conducted in submillimetric and millimetric interelectrode distances. It was deliberately done for the sake of detailed mechanistic study on the electro-precipitation of Mg(OH)<sub>2</sub> and CaCO<sub>3</sub> without the interference of external parameters.

In this chapter, the influence of additional compounds is the main topic of discussion. Understanding the influence of complex matrices of electrolyte mimicking the reclaimed wastewater (RW) properties towards both the formation of several possible electro-precipitation and the removal efficiency of a representative biorecalcitrant organic pollutant is the motivation of the study. To achieve that purpose, bottom-up approach has been adopted. It started with a comparison between basic effluent (BE) (Ca<sup>2+</sup>, Mg<sup>2+</sup> and TIC) and intermediate effluent (IE) (Ca<sup>2+</sup>, Mg<sup>2+</sup>, HCO<sub>3</sub><sup>-</sup>/CO<sub>3</sub><sup>2-</sup>, NH<sub>4</sub><sup>+</sup> and H<sub>2</sub>PO<sub>4</sub><sup>-</sup>/HPO<sub>4</sub><sup>2-</sup>). The potential addition of electro-precipitate that could be thermodynamically favored or not during the electro-oxidation has been identified at this stage. The study continued with further complexation of IE with the addition of multi-ion species (NO<sub>3</sub>-, NO<sub>2</sub>-, Cl<sup>-</sup>, SO<sub>4</sub><sup>2-</sup>, K<sup>+</sup> and Na<sup>+</sup>) whose concentrations are characteristics of those detected in reclaimed municipal wastewater. At this stage, the role of these ionic species on electro-precipitation is investigated in the electrolytic system called saline effluent (SE). Further down, the influence of the presence of organic matter towards the occurrence of electro-precipitate is characterized by adding organics (peptone and meat extract) into the SE to complete the simulated wastewater (SW). Eventually, the difference in electro-precipitate formation during the electro-oxidation process of SE and RW is compared. Finally, to put the research outcome into the practical perspective, the presence of electro-precipitation on the degradation of tylosin as model micropollutant in RW at different applied currents was evaluated.

#### VII.2. Experimental section

In **Chapter 6**, it has been shown eventually that the formation of mineral scaling was favored under submillimetric configuration. This was attributed to the lesser degree of HER evolution as well as anodic contribution of dissolved O<sub>2</sub> production towards cathode reduction activity. However, the systematic study of electro-precipitation occurrence under a microfluidic setup was still pursued in this chapter because the characterization of mineral scaling occurring under such configuration has

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not yet been done in complex matrices. Furthermore, more data are needed regarding the analysis on the fate of N and Cl species during an electro-oxidation process under submillimetric condition. For that purpose, a  $d_{\rm elec}$  of 500 µm was chosen for this chapter. It corresponds to the microfluidic configuration where the least mineral scaling was observed. Undivided parallel-plate electrochemical cell with BDD as anode (50 cm<sup>2</sup>) and stainless steel as cathode (50 cm<sup>2</sup>) was used to perform the electro-oxidation experiments. 500 mL of the effluent to be electro-oxidized was used and the effluent was recirculated with a flow rate of 100 mL min<sup>-1</sup> such as in **Chapter 6**. The applied  $j_{\rm app}$  was 4 mA cm<sup>-2</sup> throughout the series of experiment except for **Section VII.3.5** where different applied currents were intended: 0.8 and 16 mA cm<sup>-2</sup> were added.

Analytical methods have been described in detail in Chapter 3. Cation concentrations ( $Mg^{2+}$  and  $Ca^{2+}$ , supposedly identical to Mg and Ca element) were measured using ICP-OES, anion concentrations ( $NO_3^-$ ,  $NO_2^-$ ,  $SO_4^{2-}$ ,  $Cl^-$ ,  $ClO_2^-$ ,  $ClO_3^-$  and  $ClO_4^-$ ) were measured using ionic chromatography.  $NH_4^+$  was measured with HACH spectrophotometer (Nessler method, Method 8038 ( $0.02-2.50~mg~L^{-1}~NH_3-N$ )),  $PO_4^{3-}$  was quantified using PhosVer 3 method (HACH, Method 8048 ( $0.02-2.50~mg~L^{-1}~PO_4^{3-}$ )) and free chlorine was dosed using DPD method (HACH, Method 8021 ( $0.02-2.00~mg~L^{-1}$ )). TOC, TN and TIC were analyzed using TOC-TN analyzer (Shimadzu). Tylosin concentration was measured by HPLC-UV (wavelength 290 nm).

The average characteristics of all systematic simulated effluents (BE, IE, SE and SW) as well as the RW used in this work are represented in **Table VII.1**. Prior to the experiment with RW, it was filtered with standard filter paper ( $\sim$ 10  $\mu$ m pore diameter) to remove suspended solids in the effluent.

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**Table VII.1**. Average physicochemical properties of simulated and reclaimed wastewaters. Reclaimed effluent was sampled from the outlet of Reims municipal wastewater treatment plant.

Parameter / unit	Basic effluent (BE)	Intermediate effluent (IE)	Saline effluent (SE)	Simulated wastewater (SW)	Reclaimed wastewater (RW)
pН	$7.47 \pm 0.08$	$7.41 \pm 0.04$	$7.36 \pm 0.04$	$7.50 \pm 0.07$	$7.57 \pm 0.21$
Ionic conductivity / µS cm <sup>-1</sup>	1,209 ± 21	1,220 ± 14	1,660 ± 1	$1,640 \pm 14$	1,021 ± 74
$\mathrm{NH_4}^+ / \mathrm{mg} \ \mathrm{L}^{-1}$	-	1.80	$1.9 \pm 0.1$	$2.2 \pm 0.1$	$1.6 \pm 0.5$
$NO_3^- / mg L^{-1}$	-	-	1.9	$2.0 \pm 0.2$	$2.2 \pm 1.2$
$NO_2^- / mg L^{-1}$	-	-	0.40	0.40	$0.46 \pm 0.4$
$PO_4^{3-} / mg L^{-1}$	-	1.94	$2.2 \pm 0.1$	$1.8 \pm 0.1$	$1.6 \pm 0.5$
$SO_4^{2-} / mg L^{-1}$	-	-	$403 \pm 2$	401 ± 3	66 ± 7
$Cl^- / mg L^{-1}$	-	-	124 ± 2	134 ± 1	$129 \pm 38$
$Ca^{2+}$ / mg $L^{-1}$	150	155	$157 \pm 0.1$	152 ± 3	133 ± 8
$Mg^{2+}$ / $mg~L^{-1}$	5	5.4	$5.3 \pm 0.1$	$5.2 \pm 0.1$	$4.2 \pm 0.5$
$K^+ / mg L^{-1}$	-	-	17	17	$17 \pm 1.7$
$Na^+ / mg L^{-1}$	-	-	75	75	76 ± 24
$TIC / mg-C L^{-1}$	$68 \pm 0.1$	64 ± 1	68 ± 1	64 ± 2	68 ± 4
$TN / mg-N L^{-1}$	-	1.80	4.6	$5.0 \pm 0.6$	$2.1 \pm 0.5$
TOC / mg-C L <sup>-1</sup>	-	-	-	$8.2 \pm 0.9$	$4.5 \pm 0.5$

#### VII.3. Results and discussion

# VII.3.1. Electro-precipitation in electrolyte consisting of only precipitating elements

Ca<sup>2+</sup>, Mg<sup>2+</sup>, HCO<sub>3</sub>-/CO<sub>3</sub><sup>2-</sup>, NH<sub>4</sub>+ and H<sub>2</sub>PO<sub>4</sub>-/HPO<sub>4</sub><sup>2-</sup> could undergo electro-precipitation, either in simulated or real wastewater. Their plausible precipitations under the applied operating conditions are discussed in this section.

In slightly alkaline initial pH environment, which is the case of the present experimental conditions (pH = 7-8), bicarbonates (HCO<sub>3</sub><sup>-</sup>) and hydrogenophosphate (HPO<sub>4</sub><sup>2</sup>-) were the dominant forms of inorganic carbon and phosphorus respectively, following the value of pK<sub>A</sub> of the acid-base couples written in Eqs. (VII.1)-(VII.2) [1, 2]:

$$K_{A1} = \frac{[H^+][HCO_3^-]}{[H_2CO_3]}, pK_{A1} = 6.3$$
 (VII.1)

$$K_{A2} = \frac{[H^+][HPO_4^{2-}]}{[H_2PO_4^{-}]}, \text{pK}_{A2} = 7.2$$
 (VII.2)

However, due to high local pH induced on cathode surface during electrolysis owing to the reduction reactions of dissolved  $O_2$  and water (Eqs. (II.9)-(II.10)) [3-5], carbonate ( $CO_3^{2-}$ ) and phosphate ( $PO_4^{3-}$ ) ions can instantaneously become the dominant species at cathode surface, in agreement with Eqs. (VII.3)-(VII.4) following their respective pK<sub>A</sub> values [1, 2].

$$K_{A3} = \frac{[H^+][CO_3^{2-}]}{[HCO_3^-]}, \text{pK}_{A3} = 10.4$$
 (VII.3)

$$K_{A4} = \frac{[H^+][PO_4^{3-}]}{[HPO_4^{2-}]}, \text{ pK}_{A4} = 12.4$$
 (VII.4)

The evolution of concentration of ionic species during the electro-oxidation of BE and IE is given in **Fig. VII.1**. According to **Figs. VII.1(a)-VII.1(c)**, Mg<sup>2+</sup> and Ca<sup>2+</sup> concentrations decreased similarly regardless of the presence of PO<sub>4</sub><sup>3-</sup> and NH<sub>4</sub><sup>+</sup> ions under our operating conditions. It indicated the formation of Mg(OH)<sub>2</sub> and CaCO<sub>3</sub> precipitates. It has been demonstrated in a previous work that MgCO<sub>3</sub> was not formed under similar experimental setup as in this present work [5]. Mg precipitated independently in the presence of carbonates, while a similar amount of Mg(OH)<sub>2</sub> was obtained in a system without carbonates [5]. This has been further confirmed by the work of Lei et al. [6] and these authors have also ruled out the formation of dolomite (MgCa(CO<sub>3</sub>)<sub>2</sub>) after having demonstrated that Ca and Mg precipitated independently of CO<sub>3</sub><sup>2-</sup> in wide range of solution pH. In **Fig. VII.1(d)**, it was observed that the NH<sub>4</sub><sup>+</sup> concentration did not decrease when it was added into the electrolyte

containing Mg<sup>2+</sup>, Ca<sup>2+</sup>, HCO<sub>3</sub>-/CO<sub>3</sub><sup>2-</sup> and H<sub>2</sub>PO<sub>4</sub>-/HPO<sub>4</sub><sup>2-</sup>. It allows deducing that struvite (MgNH<sub>4</sub>H<sub>n</sub>PO<sub>4</sub><sup>3-n</sup>) precipitation were not thermodynamically favored under current conditions. This was due to the fact that 1:1:1 molar ratio of Mg<sup>2+</sup> (0.21 mM):NH<sub>4</sub>+ (0.10 mM):PO<sub>4</sub><sup>3-</sup> (0.015 mM) was not achieved [7], across all investigated matrices. Similar behavior was documented from the work of Lei et al. [8].

Next, when H<sub>2</sub>PO<sub>4</sub><sup>-</sup>/HPO<sub>4</sub><sup>2-</sup> was added into the BE, PO<sub>4</sub><sup>3-</sup> concentration diminished (**Fig. VII.1(e)**). Since OH was involved in such conditions of electro-oxidation treatment, phosphate might react with OH to produce phosphate radicals [9]. To check on the stability of PO<sub>4</sub><sup>3-</sup> ion under the electro-oxidation environment, an electrolysis of phosphate blank solution (1.9 mg-PO<sub>4</sub><sup>3-</sup> L<sup>-1</sup> with Na<sub>2</sub>SO<sub>4</sub> as supporting electrolyte) was performed under comparable operating conditions as the electro-oxidation experiment. **Figure VII.1(f)** illustrates the evolution of PO<sub>4</sub><sup>3-</sup> concentration during the 5 h electrolysis. A slight decrease of PO<sub>4</sub><sup>3-</sup> by 13% after the electrolysis was observed. Hence, the production of phosphate radicals (**Eqs. (VII.5)-(VII.7))** [9] as well as perphosphate (P<sub>2</sub>O<sub>8</sub><sup>4-</sup>) (**Eq. (VII.8)**) [10] might happen, but did not represent the dominant PO<sub>4</sub><sup>3-</sup> degradation route.

$$PO_4^{3-} + \bullet OH \to PO_4^{\bullet 2-} + OH^-$$
 (VII.5)

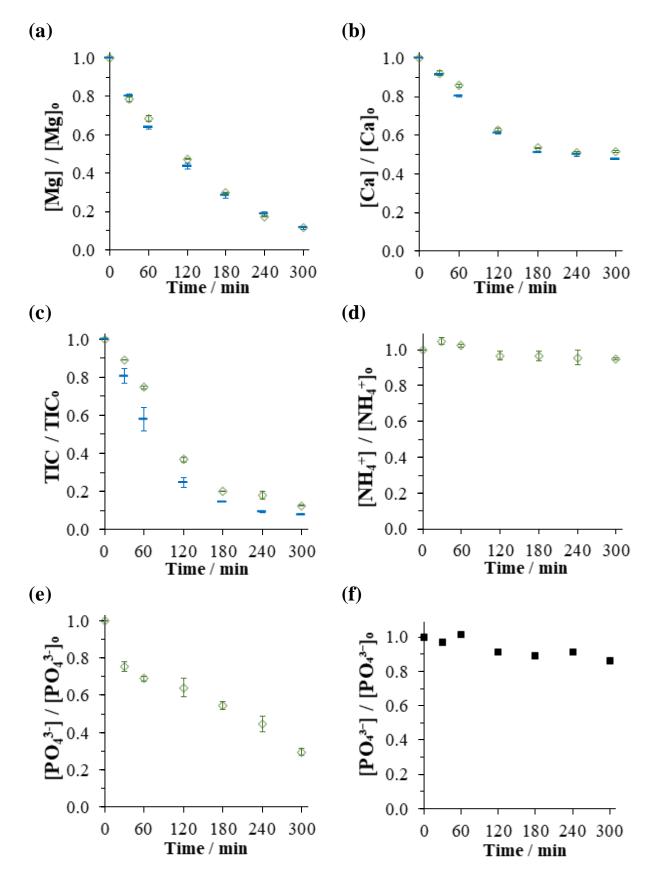
$$HPO_4^{2-} + \bullet OH \rightarrow HPO_4^{\bullet-} + OH^- \tag{VII.6}$$

$$H_2PO_4^- + \bullet OH \to H_2PO_4^{\bullet} + OH^-$$
 (VII.7)

$$2PO_4^{\bullet 2-} \to P_2O_8^{4-}$$
 (VII.8)

Another reaction mechanism could occur via electro-precipitation and could explain the decay of  $PO_4^{3-}$  noticed in **Fig. VII.1(e)**. Following the trend of  $Mg^{2+}$  concentration illustrated in **Fig. VII.1(a)**, Mg precipitated indifferently with regard to the presence or absence of phosphate. This suggests that  $Mg_3(PO_4)_2$  or its other hydrated forms were not produced. It is in agreement with the fact that the formation of magnesium phosphates was not thermodynamically feasible under present circumstances as proposed by Lei et al. **[8]**. Consequently, the decrease in  $PO_4^{3-}$  concentration was rather attributed to its precipitation with  $Ca^{2+}$  to form  $Ca_3(PO_4)_2$ . This was relevant due to the low thermodynamic solubility constant of  $Ca_3(PO_4)_2$  ( $K_{S,Ca_3(PO_4)_2} = 10^{-26}$  **[11]**), which is even lower than that of  $CaCO_3$  ( $K_{S,CaCO_3} = 10^{-9}$ ) and  $Mg(OH)_2$  ( $K_{S,Mg(OH)_2} = 10^{-12}$ ) **[5]**. Nonetheless, the concentration of phosphate was too low compared with carbonate to yield an impact on Ca precipitation, as illustrated in **Fig. VII.1(b)**. In the presence of phosphate, slightly more carbonate was present in the bulk (**Fig. VII.1(c)**), because small proportion of  $Ca^{2+}$  reacted with phosphate to produce  $Ca_3(PO_4)_2$ .

To sum up, the only feasible electro-precipitates under the present operating conditions were Mg(OH)<sub>2</sub>, CaCO<sub>3</sub> and Ca<sub>3</sub>(PO<sub>4</sub>)<sub>2</sub>. The formation of struvite, MgCO<sub>3</sub>, CaMg(CO<sub>3</sub>)<sub>2</sub> and Mg<sub>3</sub>(PO<sub>4</sub>)<sub>2</sub> were not favoured. In comparison with the literature, comparable electro-precipitate elements formed during an electrochemical process either in submillimetric (this section) or centimetric configuration was found [6, 8, 12]. This fact suggests that the anodic activity inside the micrometric reactor only contributed towards higher kinetics of mineral scaling but not on the nature of the precipitates.



**Fig. VII.1**. Concentrations of (a)  $Mg^{2+}$ , (b)  $Ca^{2+}$ , (c) TIC, (d)  $NH_4^+$  and  $PO_4^{3-}$  ((e), (f)) during the electrolysis of (—):  $Ca^{2+}$ ,  $Mg^{2+}$ ,  $HCO_3^-/CO_3^{2-}$  (BE), ( $\diamondsuit$ ):  $Ca^{2+}$ ,  $Mg^{2+}$ ,  $HCO_3^-/CO_3^{2-}$ ,  $NH_4^+$ ,  $H_2PO_4^-/HPO_4^{2-}$  (IE) and ( $\blacksquare$ ):  $PO_4^{3-}$  blank electrolytes. Anode: BDD, cathode: stainless steel and  $j_{app}$ : 4 mA cm<sup>-2</sup>.

## VII.3.2. Electro-precipitation in the presence of multi-ions representative of reclaimed wastewater

In this section, the influence of the presence of ionic species ( $NO_3^-$ ,  $NO_2^-$ ,  $SO_4^{2-}$ ,  $Cl^-$ ,  $K^+$  and  $Na^+$ ) jointly present in RW in addition to those investigated in **Section VII.3.1** (i.e.,  $Ca^{2+}$ ,  $Mg^{2+}$ ,  $HCO_3^-$ / $CO_3^{2-}$ ,  $NH_4^+$ ,  $H_2PO_4^-$ / $HPO_4^{2-}$  in IE) have been scrutinized. The added multi-ions do not form direct precipitations with the deposits possibly formed in the system but they could play indirect role towards the occurrence of electro-precipitation.

From **Fig. VII.2**, it can be observed that the addition of other ionic species into the IE did not result in significant difference under the applied conditions. From **Figs. VII.2(a)-VII.2(d)**,  $Mg^{2+}$ ,  $Ca^{2+}$ , TIC and  $PO_4^{3-}$  precipitated by  $88.2 \pm 0.3$ ,  $48.5 \pm 1.2$ ,  $87.5 \pm 0.6$  and  $70.4 \pm 4.1\%$  respectively in IE compared to 90.6,  $51.0 \pm 0.4$ ,  $89.9 \pm 0.1$  and  $72.3 \pm 6.4\%$  in SE. In addition,  $NO_3^-$  concentration decreased from  $1.9 \text{ mg L}^{-1}$  to  $1 \text{ mg L}^{-1}$  after 1 h of electrolysis and it remained constant until the end of the process. The decrease suggested that the reduction of  $NO_3^-$  to  $NO_2^-$ ,  $NH_4^+$  and  $N_2$  (**Eqs. (VII.9)-(VII.11))** [13-16] was more dominant relatively to its production (**Eqs. (VII.12)-(VII.14))** [13-16].

$$NO_3^- + 10H^+ + 8e^- \rightarrow NH_4^+ + 3H_2O$$
  $E^0 = -0.12 \text{ V/SHE}$  (VII.9)

$$NO_3^- + 6H^+ + 5e^- \rightarrow \frac{1}{2}N_2 + 3H_2O$$
  $E^0 = 1.17 \text{ V/SHE}$  (VII.10)

$$NO_3^- + 2H^+ + 2e^- \rightarrow NO_2^- + H_2O$$
  $E^0 = 0.01 \text{ V/SHE}$  (VII.11)

$$NO_2^- + H_2O \rightarrow NO_3^- + 2H^+ + 2e^-$$
 (VII.12)

$$NH_4^+ + 3H_2O \rightarrow NO_3^- + 10H^+ + 8e^-$$
 (VII.13)

$$\frac{1}{2}N_2 + 3H_2O \to NO_3^- + 6H^+ + 5e^- \tag{VII.14}$$

Furthermore, under anodic oxidation conditions, the physisorbed OH could also play its role via Eqs. (VII.15)-(VII.17) [14]:

$$NH_4^+ + \bullet OH + 3H_2O \leftrightarrow NO_3^- + 10H^+ + 8e^-$$
 (VII.15)

$$NO_2^- + \bullet OH + H_2O \leftrightarrow NO_3^- + 2H^+ + 2e^-$$
 (VII.16)

$$NH_4^+ + 3 \cdot OH + OH^- \leftrightarrow \frac{1}{2}N_2 + 4H_2O + 3e^-$$
 (VII.17)

Nonetheless, the reduction of NO<sub>3</sub><sup>-</sup> was not complete since it remained about 55% in the effluent. It means an equilibrium was reached between its rate of consumption and production.

 $NO_2^-$  initially present (0.40 mg L<sup>-1</sup>) was not detected over the course of electrolysis. It suggests that **Eqs.** (VII.12), (VII.16), (VII.18) and (VII.19) were faster [14, 17] than **Eqs.** (VII.11), (VII.20) and (VII.21), which resulted in non-accumulation of it in the bulk. Furthermore,  $NO_2^-$  was initially relatively lower than  $NO_3^-$  in the bulk.

$$NO_2^- + 4H^+ + 3e^- \rightarrow \frac{1}{2}N_2 + 2H_2O$$
 (VII.18)

$$NO_2^- + 8H^+ + 6e^- \rightarrow NH_4^+ + 3H_2O$$
 (VII.19)

$$\frac{1}{2}N_2 + 2H_2O \to NO_2^- + 4H^+ + 3e^- \tag{VII.20}$$

$$NH_4^+ + 3H_2O \rightarrow NO_2^- + 8H^+ + 6e^-$$
 (VII.21)

In IE, the concentration of  $NH_4^+$  decreased negligibly (5.0  $\pm$  0.9%) (**Fig. VII.2(e)**). Hence the reduction of  $NH_4^+$  into  $NO_3^-$  (**Eqs. (VII.13**) and (**VII.15**)), into  $NO_2^-$  (**Eq. (VII.21**)) as well as direct oxidation of  $NH_4^+$  into  $N_2$  (**Eqs. (VII.17**) and (**VII.22**)) were insignificant under our investigated conditions.

$$NH_4^+ \to \frac{1}{2}N_2 + 4H^+ + 3e^-$$
 (VII.22)

The slight  $NH_4^+$  increase in the first hour of electrolysis could be ascribed to  $NO_3^-$  reduction (**Eq.** (VII.9)).

Contrastingly, the evolution of  $NH_4^+$  concentration changed dramatically in multi-ions electrolyte (SE), as can be reckoned from **Fig. VII.2(e)**, where 99.0  $\pm$  4.8%  $NH_4^+$  was removed. This quasi-complete removal of  $NH_4^+$  was associated with the presence of  $Cl^-$ : there was a fast reaction between  $NH_4^+$  with HClO producing inorganic chloramines [15, 17-21] as listed in **Eqs. (VII.23)-(VII.29)**.

$$2Cl^{-} \rightarrow Cl_{2} + 2e^{-}$$
 E<sup>0</sup> = 1.36 V/SHE (VII.23)

$$Cl_2 + H_2O \to HClO + H^+ + Cl^-$$
 (VII.24)

$$\frac{1}{2}Cl_2 + H_2O \rightarrow HClO + H^+ + e^-$$
 E<sup>0</sup> = 1.61 V/SHE (VII.25)

$$HClO + NH_4^+ \rightarrow NH_2Cl + H_2O + H^+$$
 (VII.26)

$$NH_2Cl + HClO \rightarrow NHCl_2 + H_2O$$
 (VII.27)

$$NH_2Cl + HClO \rightarrow NCl_3 + H_2O$$
 (VII.28)

$$NH_2Cl + H^+ \to HClO + NH_4^+ \tag{VII.29}$$

This deduction can be justified by the oxidation of Cl<sup>-</sup> to Cl<sub>2</sub> (**Eq.** (**VII.23**)) and fast conversion of Cl<sub>2</sub> to hypochloric acid (HClO) (**Eqs.** (**VII.24**)-(**VII.25**)), which led to the decrease of Cl<sup>-</sup> concentration in **Fig. VII.3(a)** and increasing trend of HClO in **Fig. VII.3(b)**. The quick conversion from Cl<sup>-</sup> into Cl<sub>2</sub> and HClO was relevant given higher applied anode potential ( $E_A = 2.40 \text{ V/Ag-AgCl}$  or 2.60 V/SHE) relatively to  $E_{\text{Cl}_2/\text{Cl}^-}^0$  (1.36 V/SHE) (**Eq.** (**VII.23**)) and  $E_{\text{HClO/Cl}_2}^0$  (1.61 V/SHE) (**Eq.** (**VII.25**)). HClO further reacted with NH<sub>4</sub><sup>+</sup> producing NHCl<sub>2</sub>, NH<sub>2</sub>Cl and NCl<sub>3</sub> (**Eqs.** (**VII.26**)-(**VII.28**)). Moreover, NH<sub>4</sub><sup>+</sup> reacted with HClO not only to form inorganic chloramines, but also to generate NO<sub>3</sub><sup>-</sup> and N<sub>2</sub> as shown in **Eqs.** (**VII.30**)-(**VII.31**).

$$NH_4^+ + 4HClO \rightarrow NO_3^- + H_2O + 6H^+ + 4Cl^-$$
 (VII.30)

$$2NH_4^+ + 3HClO \rightarrow N_2 + 3H_2O + 5H^+ + 3Cl^-$$
 (VII.31)

In excess of chlorides with respect to molar ratio of  $NH_4^+$  ( $Cl^-/NH_4^+ > 7.6$ ), the so-called break point can be reached. In our condition, the ratio  $N_{Cl^-}/N_{NH_4^+}$  was higher than 25, so the produced inorganic chloramines in effluent bulk could be further oxidized to form gaseous  $N_2$  and nitrogen oxides ( $NO_X$ ) (Eqs. (VII.32)-(VII.33)) [18, 19]. The group of  $N_2$  and  $NO_X$  has been called volatile N (N-volatile) [13, 22]. The formation of N-volatile while treating N-species in multi-ions electrolyte was confirmed by the drop of TN concentration in Fig. VII.2(f).

$$4NH_2Cl + 3Cl_2 + H_2O \rightarrow N_2 + N_2O + 10HCl$$
 (VII.32)

$$NH_2Cl + NHCl_2 \rightarrow N_2 + 3HCl$$
 (VII.33)

$$NH_2Cl + NHCl_2 + HClO \rightarrow N_2O + 4HCl$$
 (VII.34)

$$2NH_2Cl + HClO \rightarrow N_2O + 3HCl \tag{VII.35}$$

The intermediary behavior of the production of HClO was not only due to the reactions with N-species, but also to its conversion back into  $Cl_2$  (**Eq.** (**VII.36**)). Furthermore, its direct reaction into chlorate ( $ClO_3^-$ ) (**Eqs.** (**VII.37**)-(**VII.39**)) [10, 17] as well as by intermediary of chlorite ( $ClO_2^-$ ) (**Eqs.** (**VII.40**)-(**VII.41**)) [10] have commonly been documented. This was proven by the increasing concentration of  $ClO_3^-$  over the course of electrolysis depicted in **Fig. VII.3(c)**. Not only was the chemical conversion of HClO to  $ClO_3^-$  quick (**Eqs.** (**VII.38**)-(**VII.39**)), its standard potential of reduction is also low  $E^0_{HClO/ClO_3^-}$  (0.46 V/SHE) (**Eq.** (**VII.37**)). The conversion to  $ClO_3^-$  in presence of HClO was highly favorable.

$$HClO + H^{+} + e^{-} \rightarrow \frac{1}{2}Cl_{2} + H_{2}O$$
 (VII.36)

$$6HClO + 3H_2O \rightarrow 2ClO_3^- + 4Cl^- + \frac{3}{2}O_2 + 12H^+ + 6e^- \qquad E^0 = 0.46 \text{ V/SHE}$$
 (VII.37)

$$HClO + 2ClO_2 + H_2O \rightarrow 2ClO_3^- + HCl + 2H^+$$
 (VII.38)

$$HClO \rightarrow ClO_3^- + HCl + Cl^-$$
 (VII.39)

$$HClO + H_2O \rightarrow ClO_2^- + 3H^+ + 2e^-$$
 (VII.40)

$$ClO_2^- + 2H_2O \rightarrow ClO_3^- + 2H^+ + 2e^-$$
 E<sup>0</sup> = 0.33 V/SHE (VII.41)

Furthermore, ClO<sub>3</sub><sup>-</sup> can also be produced directly from Cl<sup>-</sup> and Cl<sub>2</sub> as shown in **Eqs.** (VII.42)-(VII.44) [10]. The applied  $E_A$  (2.60 V/SHE) was relatively larger than both the  $E_{Cl_2/ClO_3}^0$  (1.45 V/SHE) (**Eq.** (VII.42)) and  $E_{HClO/ClO_3}^0$  (1.47 V/SHE) (**Eq.** (VII.43)).

$$Cl^{-} + 3H_2O \rightarrow ClO_3^{-} + 6H^{+} + 6e^{-}$$
  $E^0 = 1.45 \text{ V/SHE}$  (VII.42)

$$\frac{1}{2}Cl_2 + 3H_2O \rightarrow ClO_3^- + 6H^+ + 5e^-$$
  $E^0 = 1.47 \text{ V/SHE}$  (VII.43)

$$3Cl_2 + 3H_2O \rightarrow ClO_3^- + 5Cl^- + 6H^+$$
 (VII.44)

Perchlorates ( $ClO_4^-$ ) were also produced in the treated effluent as can be observed in **Fig. VII.3(d)**. The generation of  $ClO_4^-$  has been well documented when BDD was used as anode [16, 23, 24]. The

oxidation of  $\text{ClO}_3^-$  led to  $\text{ClO}_4^-$  as written in **Eq. (VII.45)**. This conversion was thermodynamically favorable given by the value of  $E^0_{\text{ClO}_4^-/\text{ClO}_3^-}$  (1.20 V/SHE) which was lower than the applied  $E_A$  (2.60 V/SHE). This explains progressively slower production rate of  $\text{ClO}_3^-$  (**Fig. VII.3(c**)) as a function of electrolysis time, in contrast with the increasing production rate of  $\text{ClO}_4^-$  (**Fig. VII.3(d**)).

$$ClO_3^- + H_2O \rightarrow ClO_4^- + 2H^+ + 2e^-$$
 E<sup>0</sup> = 1.20 V/SHE (VII.45)

The accumulation of ClO<sub>3</sub><sup>-</sup> and ClO<sub>4</sub><sup>-</sup> indicated that their decomposition rates corresponding to **Eqs.** (VII.46)-(VII.50) were much slower in comparison to their production rate.

$$ClO_3^- + 5Cl^- + 6H^+ \rightarrow 3Cl_2 + 3H_2O$$
 (VII.46)

$$ClO_3^- + 6H^+ + 5e^- \rightarrow \frac{1}{2}Cl_2 + 3H_2O$$
 (VII.47)

$$ClO_3^- + 2H^+ + 2e^- \rightarrow ClO_2^- + 3H_2O$$
 (VII.48)

$$2ClO_3^- + 4Cl^- + \frac{3}{2}O_2 + 12H^+ + 6e^- \rightarrow 6HClO + 3H_2O$$
 (VII.49)

$$ClO_3^- + 6H^+ + 6e^- \rightarrow Cl^- + 3H_2O$$
 (VII.50)

The accumulation of  $ClO_2^-$  remained below the detection limit throughout the electrolysis owing to its fast oxidation rate to  $ClO_3^-$  (Eq. (VII.41)) and reduction to HClO (Eq. (VII.51)) [10, 16].

$$ClO_2^- + 3H^+ + 2e^- \to HClO + H_2O$$
 (VII.51)

In addition, the decrease of Cl<sup>-</sup> (**Fig. VII.3(a**)) was partly due to the concurrent conversion towards volatile Cl (Cl-volatile), which is ascribed as Cl<sub>2</sub>, ClO<sub>2</sub> and Cl<sub>2</sub>O following **Eqs. (VII.52**)-(**VII.59**) [10, 25]:

$$Cl^{-} + 2H_2O \rightarrow ClO_2 + 4H^{+} + 5e^{-}$$
 E<sup>0</sup> = 1.60 V/SHE (VII.52)

$$Cl_2 + 4H_2O \rightarrow 2ClO_2 + 8e^-$$
 E<sup>0</sup> = 1.54 V/SHE (VII.53)

$$\frac{1}{2}Cl_2 + 2H_2O \rightarrow HClO_2 + 3H^+ + 3e^-$$
 E<sup>0</sup> = 1.63 V/SHE (VII.54)

$$2Cl^{-} + H_2O \rightarrow Cl_2O + 2H^{+} + 4e^{-}$$
 (VII.55)

$$HClO + H_2O \rightarrow HClO_2 + 2H^+ + 2e^-$$
 E<sup>0</sup> = 1.68 V/SHE (VII.56)

$$HClO_2 \rightarrow ClO_2 + H^+ + e^-$$
 E<sup>0</sup> = 1.28 V/SHE (VII.57)

$$ClO_3^- + H_2O + e^- \rightarrow ClO_2 + 2OH^-$$
 E<sup>0</sup> = 1.17 V/SHE (VII.58)

$$ClO_2^- \rightarrow ClO_2 + e^-$$
 E<sup>0</sup> = 0.95 V/SHE (VII.59)

Their occurrence was electrochemically feasible since the applied  $E_A$  (2.60 V/SHE) was superior to all redox couples of Cl-volatile species listed in **Eqs.** (VII.52)-(VII.59).

Moreover, as it was the case with N-species, Cl-species that diffused close to BDD surface would react with OH (Eqs. (VII.60)-(VII.68)) [10, 19]:

$$HClO + \bullet OH \rightarrow \bullet ClO + H_2O$$
 (VII.60)

$$\bullet ClO + \bullet OH \to ClO_2^- + H^+ \tag{VII.61}$$

$$ClO_2^- + \bullet OH \rightarrow \bullet ClO_2$$
 (VII.62)

$$\bullet ClO_2 + \bullet OH \to ClO_3^- + H^+ \tag{VII.63}$$

$$H_2O \to \bullet OH + H^+ + e^-$$
 (VII.64)

$$Cl^- + \bullet OH \rightarrow ClO^- + HClO$$
 (VII.65)

$$ClO^{-} + \bullet OH \rightarrow ClO_{2}^{-} + H^{+} + e^{-}$$
 (VII.66)

$$ClO_2^- + \bullet OH \to ClO_3^- + H^+ + e^-$$
 (VII.67)

$$ClO_3^- + \bullet OH \to ClO_4^- + H^+ + e^-$$
 (VII.68)

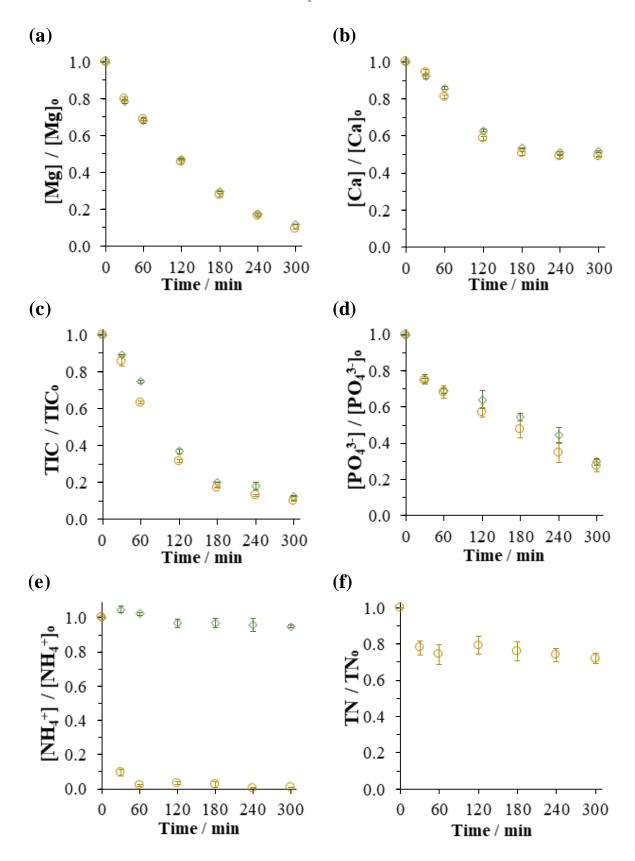
 $SO_4^{2-}$  concentrations during the electrolysis of IE as well as SW and RW are plotted in **Fig. VII.3(e)**. They remained constant throughout the process. This proved that negligible amount of  $SO_4^{2-}$  was decomposed to peroxodisulfate ( $S_2O_8^{2-}$ ) and sulfate radical ( $SO_4^{\bullet-}$ ) according to **Eqs. (VII.69**)-(VII.71) [5, 26, 27] under all investigated conditions throughout this chapter.

$$2SO_4^{2-} \rightarrow S_2O_8^{2-} + 2e^-$$
 (VII.69)

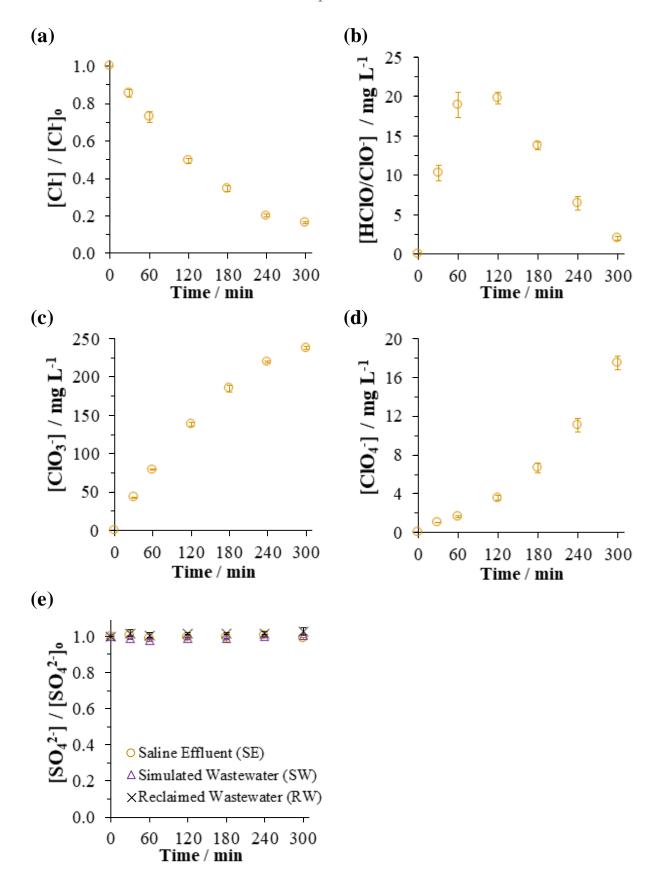
•
$$0H + 2SO_4^{2-} \rightarrow SO_4^{\bullet-} + OH^-$$
 (VII.70)

$$SO_4^{\bullet -} + SO_4^{\bullet -} \to S_2O_8^{2-}$$
 (VII.71)

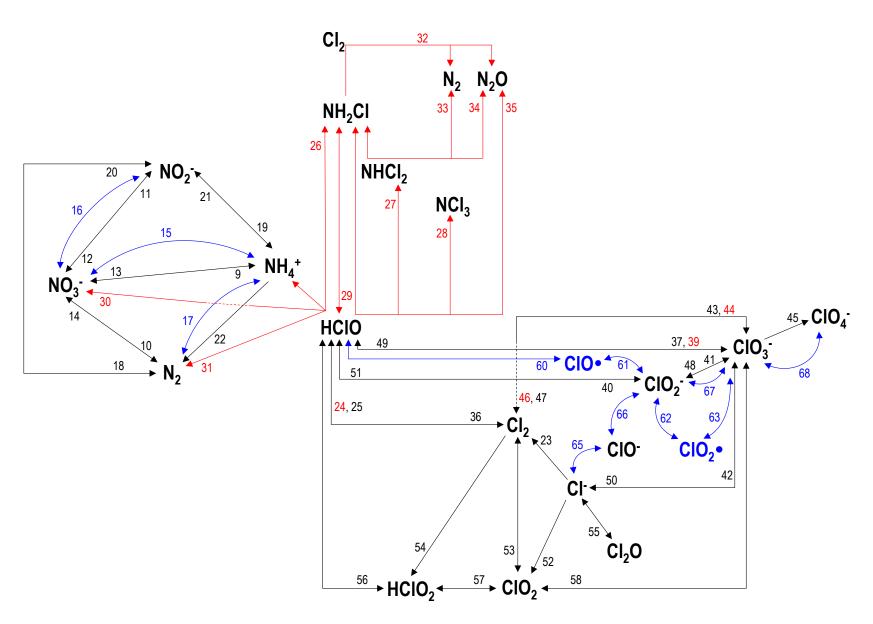
To summarize, the presence of multi-ions whose concentrations were representative to those present in RW yielded insignificant influence towards the quantity of Mg(OH)<sub>2</sub>, CaCO<sub>3</sub> and Ca<sub>3</sub>(PO<sub>4</sub>)<sub>2</sub> precipitations under simulated environment. However, the fate of other inorganic species during the electro-oxidation process could be followed. A mechanistic scheme involving the interrelation of the proposed half-equations of N and Cl species together with their reactions with \*OH has been proposed in Fig. VII.4. In the absence of Cl<sup>-</sup>, the oxidation of NH<sub>4</sub><sup>+</sup> into NO<sub>3</sub><sup>-</sup> and N-volatile was inefficient since 95% of NH<sub>4</sub><sup>+</sup> (1.7 mg-NH<sub>4</sub><sup>+</sup> L<sup>-1</sup>) remained in the electrolyte. It suggested that the physisorbed \*OH was not effective to oxidize NH<sub>4</sub><sup>+</sup>. In the presence of Cl<sup>-</sup>, mediated oxidant HClO played central role towards the degradation of NH<sub>4</sub><sup>+</sup> principally into dissolved chloramines as well as N-volatile. The interrelation of reduction and oxidation reactions involving radicals yielded ClO<sub>3</sub><sup>-</sup> and ClO<sub>4</sub><sup>-</sup> as principal chlorinated byproducts. Despite the presence of Cl<sup>-</sup>, intricate redox reactions including radical activities of N species only resulted in partial reduction of NO<sub>3</sub><sup>-</sup> into NH<sub>4</sub><sup>+</sup> and N-volatile. Total denitrification did not occur in the present experimented reactor design since it remained 55% of NO<sub>3</sub><sup>-</sup> (i.e., 1 mg-NO<sub>3</sub><sup>-</sup> L<sup>-1</sup>) in the treated media.



**Fig. VII.2**. Concentrations of (a)  $Mg^{2+}$ , (b)  $Ca^{2+}$ , (c) TIC, (d)  $PO_4^{3-}$ , (e)  $NH_4^+$  and (f) TN during the electrolysis of ( $\diamondsuit$ ):  $Ca^{2+}$ ,  $Mg^{2+}$ ,  $HCO_3^-/CO_3^{2-}$ ,  $NH_4^+$ ,  $H_2PO_4^-/HPO_4^{2-}$  (IE) and ( $\bigcirc$ ): SE (without organics) electrolytes. Anode: BDD, cathode: stainless steel and  $j_{app}$ : 4 mA cm<sup>-2</sup>.



**Fig. VII.3**. Concentrations of (a) Cl<sup>-</sup>, (b) HClO/ClO<sup>-</sup>, (c) ClO<sub>3</sub><sup>-</sup>, (d) ClO<sub>4</sub><sup>-</sup> and (e) SO<sub>4</sub><sup>2-</sup> during the electrolysis of SE. Anode: BDD, cathode: stainless steel and  $j_{app}$ : 4 mA cm<sup>-2</sup>.

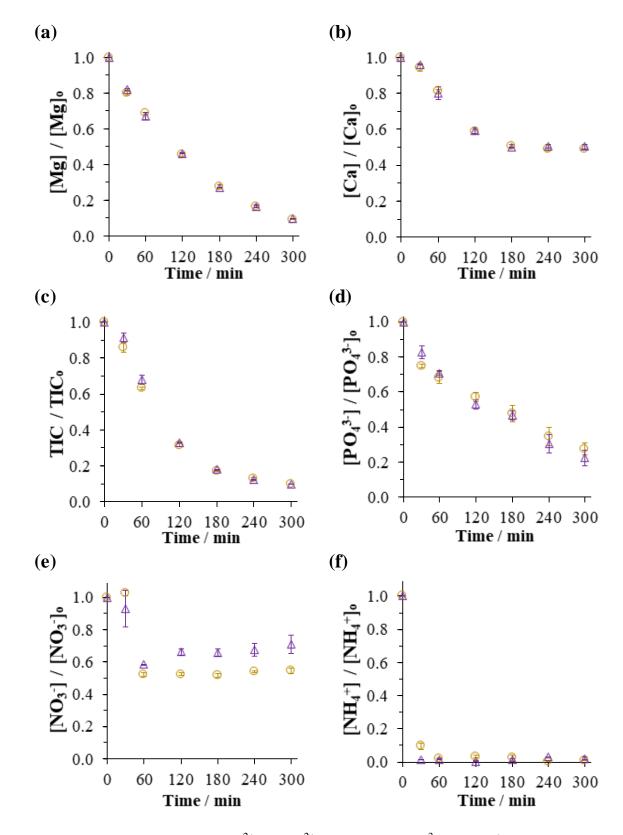


**Fig. VII.4**. Mechanistic scheme of the evolution of N and Cl species during the electro-oxidation process. The numbers are referred to the equations cited in the text. **Black**: direct electron transfer, **blue**: reactions involving radicals and **red**: chemical reactions and/or hydrolysis.

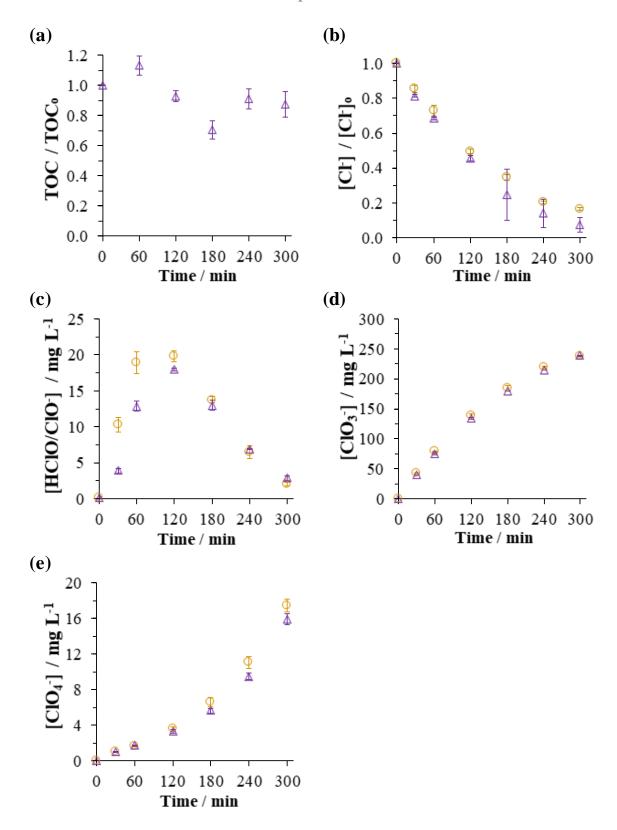
## VII.3.3. Electro-precipitation in the presence of organic matter in simulated wastewater

The organic content of RW was simulated by adding 32 mg L<sup>-1</sup> of peptone and 22 mg L<sup>-1</sup> of meat extract following the procedure carried out elsewhere [28, 29]. The difference in the evolution of ionic and organic species during an electro-oxidation process of simulated electrolyte in the absence and presence of organics can be found in Figs. VII.5 and VII.6. In overall, by adding organic matter into the simulated inorganic effluent, no significant effect on the occurrence of electro-precipitation was observed under these applied conditions. The difference between the precipitation of Mg, Ca, TIC and PO<sub>4</sub><sup>3-</sup> in Figs. VII.5(a)-VII.5(d) was only about 0.5, 1.3, 0.2 and 4.8 % respectively. This trend could be due to high local pH [5] at the proximity of cathode that could lift the inhibitory effect of organics including humic acids [30, 31]. Our results corroborated with those documented by Lei et al. in which they did not observe any inhibiting effect of humic acids on calcium phosphate deposition under macro-reactor [12, 32]. At an appropriate humic acid concentration, it rather formed complexes with calcium and phosphate to favor calcium phosphate precipitation [12, 32]. Figure VII.5(e) depicts the evolution of NO<sub>3</sub> during the electro-oxidation of SE and SW. It can be noticed that less NO<sub>3</sub> was reduced to NH<sub>4</sub> under current applied conditions. This observation could be attributed to the presence of organics in the electrolyte. Due to competition for electronic exchanges, less NO<sub>3</sub><sup>-</sup> was converted to NH<sub>4</sub><sup>+</sup>. From Fig. VII.5(f), NH<sub>4</sub><sup>+</sup> concentration was observed to already disappear after 0.5 h of electrolysis. As discussed in Section VII.3.2, in the presence of excess Cl<sup>-</sup> in the effluent, NH<sub>4</sub><sup>+</sup> instantaneously reacted with mediated chlorinated oxidants to yield chloramines, N<sub>2</sub> and NO<sub>2</sub> (Eqs. (VII.26)-(VII.35)). Furthermore, in the presence of organics, organochloride species might as well be produced as a result of reactions between organics with mediated oxidants and OH [22, 33]. The evolution of TOC plotted in Fig. VII.6(a) could support the aforementioned reasoning whereby the TOC dropped by about 14% during the electro-oxidation treatment. In conjunction with the decrease of organic content (Fig. VII.6(a)), the concentration of Cl<sup>-</sup> decreased more in the simulated effluent containing organics compared to the one without organics. This suggested that there was an additional consumption of Cl-mediated oxidants in the presence of organic matter. The hypothesis could be further affirmed by the plot of HClO/ClO in Fig. VII.6(c). The HClO/ClO concentration was lower in the system containing organics as a consequence of its reaction with not only NH<sub>4</sub><sup>+</sup> but additionally organic matter. Eventually, slightly less Cl<sup>-</sup>, ClO<sup>-</sup> and ClO<sub>2</sub> were available for their conversion towards ClO<sub>3</sub> and ClO<sub>4</sub>, which explained lower ClO<sub>3</sub> and ClO<sub>4</sub><sup>-</sup> transformation depicted in Figs. VII.6(d) and VII.6(e).

To conclude, the presence of organics contributed to trivial difference in the occurrence of electro-precipitation during the electro-oxidation treatment of simulated effluent. It has been shown here that the inhibition of cathodic deposition by organics was not present due to high local pH. Comparable results have been obtained under both microfluidic and macrometric setups.



**Fig. VII.5**. Concentrations of (a)  $Mg^{2+}$ , (b)  $Ca^{2+}$ , (c) TIC, (d)  $PO_4^{3-}$ , (e)  $NH_4^+$  and (f)  $NO_3^-$  during the electrolysis of ( $\bigcirc$ ): SE and ( $\triangle$ ): SW. Anode: BDD, cathode: stainless steel and  $j_{app}$ : 4 mA cm<sup>-2</sup>.



**Fig. VII.6**. Concentrations of (a) TOC, (b) Cl<sup>-</sup>, (c) HClO/ClO<sup>-</sup>, (d) ClO<sub>3</sub><sup>-</sup> and (e) ClO<sub>4</sub><sup>-</sup> during the electrolysis of ( $\bigcirc$ ): SE and ( $\triangle$ ): SW. Anode: BDD, cathode: stainless steel and  $j_{app}$ : 4 mA cm<sup>-2</sup>.

### VII.3.4. Electro-precipitation in simulated versus reclaimed wastewater effluent

The formation of electro-precipitates during the electro-oxidation of simulated and real effluent is compared in Fig. VII.7. It can be noticed from Figs. VII.7(a) and VII.7(b) that more precipitation of Mg and Ca were obtained in the matrix of real effluent in comparison with the simulated counterpart. Mg and Ca precipitated by 92.7  $\pm$  3.1% and 70.0  $\pm$  0.4% respectively in real effluent, whilst they precipitated by  $90.1 \pm 0.1\%$  and  $49.7 \pm 2.7\%$  in simulated effluent respectively. Two possible reasons could be given to justify the observed trend. Firstly, though best effort has been done to achieve close characteristics in overall aspects between both effluents (Table VII.1), there could have been in reality more organic substances that could co-precipitate with Mg<sup>2+</sup> and/or Ca<sup>2+</sup> on electrode surfaces which would reduce the cations in the bulk. Fig. VII.8 illustrates the fluorescence spectra of both the simulated and real RW. It clearly shows a significant fluorescence response with RW, from 290 to 360 nm excitation wavelength bands (raw intensity of 424 to 352), in comparison to the response given by SW with low intensity peaks of 43 and 61 at 285 and 359 nm, respectively. Higher fluorescence response in RW was due to its higher fluorophore contents, which should be composed of several combined aromatic and/or cyclic groups with  $\pi$ -bonds. These are the characteristics of colloidal and dissolved organic matter in the RW which are recalcitrant since they are the residue in the effluent of WWTPs [34]. This proved that RW effluent contained more complex matrices of dissolved organics compared to the simulated effluent. They could form complexes with  $Ca^{2+}$  and/or Mg<sup>2+</sup> at the proximity of electrodes or even undergo co-precipitation with Ca and Mg on cathode surface, as suggested by other authors [12, 32, 35]. Secondly, there was about six-time higher concentration of SO<sub>4</sub><sup>2-</sup> between the simulated and real wastewater effluent. Higher SO<sub>4</sub><sup>2-</sup> content in the SW was unavoidable, since it was the counter ion of several salts used to prepare the effluent including the sulfuric acid to slightly adjust effluent pH to a comparable value. SO<sub>4</sub><sup>2-</sup> was selected since it was inactive with regards to anodic decomposition in the applied current condition as well as to the precipitating reactions of electrodeposition (Section VII.3.2 and Reference [5]). Nonetheless, the excess presence of  $SO_4^{2-}$  might result in a passive inhibiting effect towards the formation of electro-precipitation as demonstrated in literature. Some authors who found similar finding suggested that the impedance of CaCO<sub>3</sub> growth occurred due to: (1) lower cathodic local pH which comes with increasing SO<sub>4</sub><sup>2-</sup> concentration [36, 37], (2) adsorption/incorporation of SO<sub>4</sub><sup>2-</sup> on CaCO<sub>3</sub> crystals [38] or (3) higher solubility of CaCO<sub>3</sub> in the presence of SO<sub>4</sub><sup>2-</sup> [39, 40].

Then, as the Ca<sup>2+</sup> precipitated more in the real effluent, it would correspondingly be expected that more TIC in the bulk decreased. However, it was not the case according to the evolution of bulk TIC depicted in **Fig. VII.7(c)**. Surely, more CaCO<sub>3</sub> was formed on cathode given relatively lower concentration of PO<sub>4</sub><sup>3-</sup> versus TIC (1.6 mg-PO<sub>4</sub><sup>3-</sup> L<sup>-1</sup> vs. 68 mg-C L<sup>-1</sup>) in the effluent. Consequently,

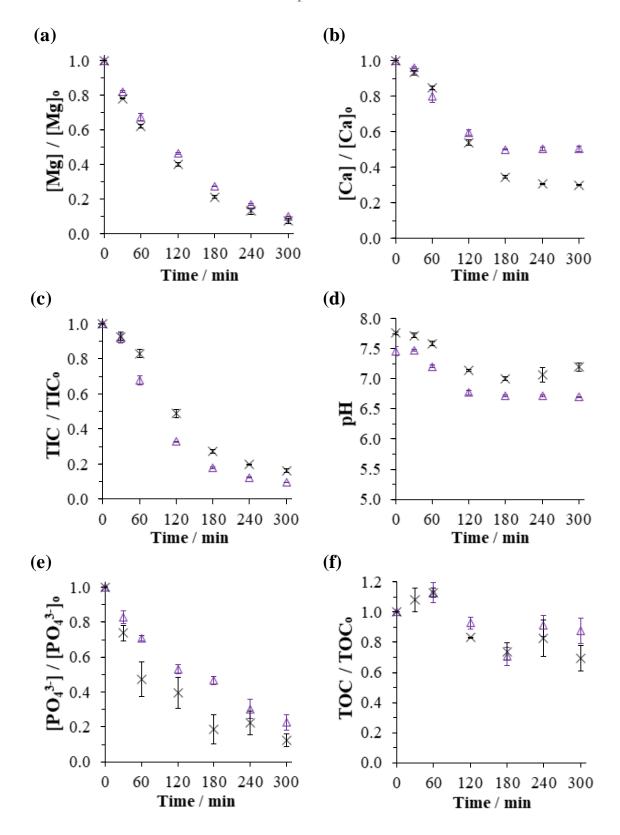
the precipitation of phosphate only took a small portion of free Ca<sup>2+</sup>. Lower TIC in the bulk could then be explained by higher buffering capacity in the real effluent compared to simulated effluent. It can be justified by the evolution of bulk pH provided in **Fig. VII.7(d)**. The trend of PO<sub>4</sub><sup>3-</sup> evolution is plotted in **Fig. VII.7(e)** and it corroborated with the higher Ca precipitation which led to higher precipitation with phosphate. Then, according to **Fig. VII.7(f)**, it can be seen that higher mineralization occurred in real matrix effluent as compared to the simulated effluent (28.3% vs. 14.4%), which indicated that more organic compound was degraded in the former.

**Figure VII.9** presents the percentage of distributions of N species and Cl species during the electro-oxidation of simulated and real effluent experiments. The evolutions of NO<sub>3</sub><sup>-</sup> and NH<sub>4</sub><sup>+</sup> species observed previously remained valid; in both effluents, there was a slight decrease in NO<sub>3</sub><sup>-</sup> before it remained practically constant till the end of electrolysis. NH<sub>4</sub><sup>+</sup> concentration decreased abruptly from the beginning of electrolysis using both effluents, also following the trend deduced in previous sections. N-volatile (i.e. N<sub>2</sub> and NO<sub>x</sub>) was calculated from mass balance between initial TN and the evolved TN during electrolysis. From **Figs. VII.9(a)** and **VII.9(b)**, it is interesting to note that more N-volatile was produced while treating simulated effluent (34%) relatively to N-volatile generated from real effluent (< 2%). Contrastingly, the N-combined, which regroups inorganic chloramine and organochloride species, was found to be higher in real effluent (87%) than in simulated effluent (57%). It could be therefore concluded that in both simulated and real effluents, the byproduct of N-species after 5 h electro-oxidation treatment consisted of inorganic chloramines and organochloride compound, but with high N-volatile produced in simulated in comparison to the real effluent. The discussion on the distribution percentage of Cl-species would help to conclude on this trend.

**Figures VII.9(c)** and **VII.9(d)** depict the percentage of distribution of Cl-species over the course of electrolysis in simulated and real effluents, respectively. It can be noticed that 92.6 ± 8.0% of Cl<sup>-</sup> in simulated effluent was converted to soluble chlorinated byproducts and Cl-volatile via chain reactions discussed in **Section VII.3.2**, while lower Cl<sup>-</sup> conversion of 76.7 ± 4.2% was evaluated using real effluent. Cl-volatile was determined through mass balance between initial Cl content subtracted with all dissolved chlorinated species during the electro-oxidation process. Then, according to **Figs. VII.9(c)** and **VII.9(d)**, lower HClO/ClO<sup>-</sup> was measured in the simulated system (18 mg-Cl<sub>2</sub> L<sup>-1</sup>) after 2 h of electrolysis in comparison with the real system (27.5 mg-Cl<sub>2</sub> L<sup>-1</sup>). These two differences could be explained by the fate of Cl<sup>-</sup> decomposition pathway in both investigated electrolyte matrices. Less Cl<sup>-</sup> degradation by about 16% in the real effluent indicated that there were much more competitive reactions present in the system, which slowed down the kinetics of Cl<sup>-</sup> consumption. This statement could be supported by the accumulation of HClO/ClO<sup>-</sup> in the bulk, where the reactions of free chlorine with organics (in real effluent) possibly with cyclic bonds and/or aromatic compounds were

kinetically slower than the decomposition and electronic transfer of HClO/ClO<sup>-</sup> to ClO<sub>3</sub><sup>-</sup>, ClO<sub>4</sub><sup>-</sup> and Cl-volatile [33, 41]. Consequently, since the SW contained simpler organic matrix, greater conversion of Cl<sup>-</sup> into HClO/ClO<sup>-</sup> was possible, which favored the oxidation into ClO<sub>3</sub><sup>-</sup>, ClO<sub>4</sub><sup>-</sup> and Cl-volatile, measured at 238 mg L<sup>-1</sup>, 16 mg L<sup>-1</sup> and 12% of total chlorine, respectively, after 5 h of electrolysis. Due to competitive reactions of Cl<sup>-</sup> oxidation and degradation of organics by electron transfer and OH, Cl<sup>-</sup> conversion towards ClO<sub>3</sub><sup>-</sup>, ClO<sub>4</sub><sup>-</sup> and Cl-volatile in real matrix was delayed [41-43]. It led to their respective lower concentrations in the bulk, i.e., 193 mg L<sup>-1</sup>, 4.7 mg L<sup>-1</sup> and below 2% of total chlorine, after 5 h of treatment.

In summary, after 5 h of electro-oxidation treatment of simulated and real wastewater, more electro-precipitation was observed using real wastewater. This was probably due to a more complex organic matrix that could favor co-precipitation with Mg<sup>2+</sup> and Ca<sup>2+</sup> in real effluent and/or inhibiting effect of SO<sub>4</sub><sup>2-</sup> in simulated wastewater. Less decomposition of Cl<sup>-</sup> was observed in real effluent owing to concurrent organic reactions on anode surface. Cl-mediated oxidants rather reacted with organic pollutants in the bulk of reclaimed effluent, which led to a delay into the conversion of inorganic chlorinated end-products.



**Fig. VII.7**. Concentrations of (a)  $Mg^{2+}$ , (b)  $Ca^{2+}$ , (c) TIC, (d) pH, (e)  $PO_4^{3-}$  and (f) TOC during the electrolysis of ( $\triangle$ ): simulated (SW) and ( $\times$ ): real effluent (RW). Anode: BDD, cathode: stainless steel and  $j_{app}$ : 4 mA cm<sup>-2</sup>.

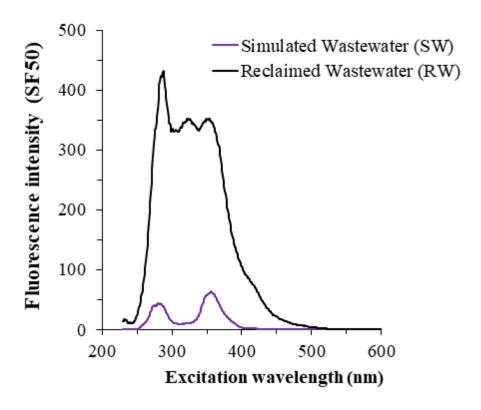
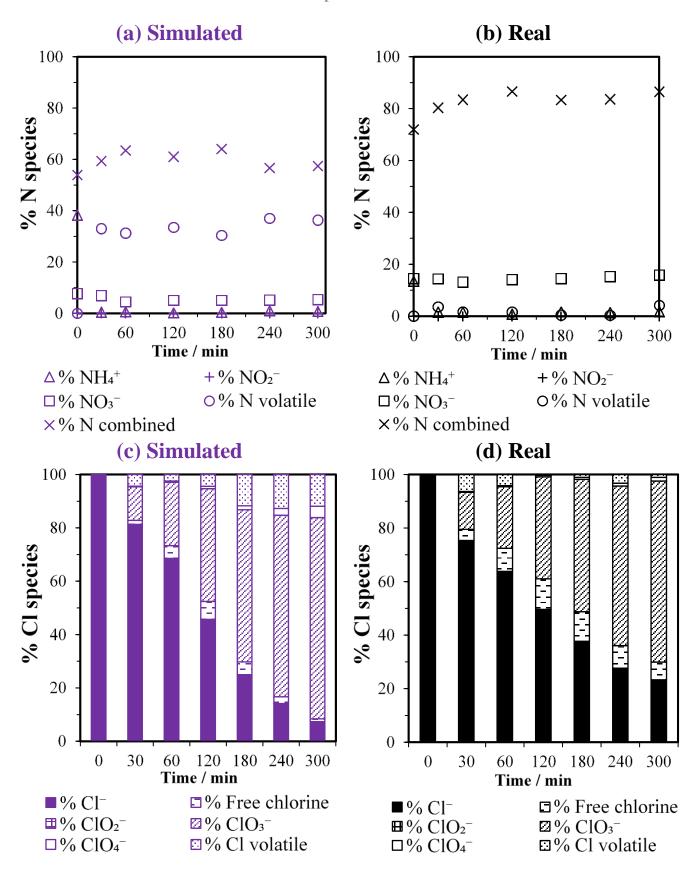


Fig. VII.8. Fluorescence spectra of simulated (SW) and reclaimed (RW) wastewater.



**Fig. VII.9**. Percentage distributions of N species in (a) simulated (SW) and (b) real effluent (RW) as well as Cl species in (c) simulated (SW) and (d) real effluent (RW) during the electro-oxidation treatment. Anode: BDD, cathode: stainless steel and  $j_{app}$ : 4 mA cm<sup>-2</sup>.

# VII.3.5. Role of electro-precipitation on the degradation of tylosin as model micropollutant in real wastewater effluent

The influence of occurring electro-precipitation on cathode surface towards the organic mineralization and tylosin degradation as model micropollutant in real wastewater effluent is investigated in this section. Tylosin was selected as model compound as it is a common pharmaceutical micropollutant found at the outlet of WWTP receiving industrial dairy and livestock farming wastes since tylosin is used as a veterinary antibiotic [44-49]. Since it is not found in typical domestic wastewater at the out stream of treatment plant, such as our sampled wastewater (Reims municipal WWTP, France), our sampled effluent was spiked with 10 mg L<sup>-1</sup> of tylosin. The spiked concentration was larger than the average concentration of pharmaceutical compounds detected in the treated waste stream ( $< 0.5 \,\mu g \, L^{-1}$  [50, 51]). However, it would help to better characterize the influence of electro-precipitation on the degradation kinetics of tylosin compound. The chosen tylosin concentration seemed appropriate because comparable concentrations of pharmaceuticals were also used elsewhere to spike RW [34, 52-54]. In section VII.3.5.1, the occurrence of electro-precipitation was investigated over the course of electro-oxidation treatment at different  $j_{app}$ . The fate of inorganic species during these processes has been evaluated. The next sub-section was devoted to discuss the influence of electro-precipitation at different investigated  $j_{app}$  on tylosin degradation and total organic mineralization.

### VII.3.5.1. Evolution of electro-precipitation at different applied current densities

The values of  $j_{app}$  were determined as function of initial dimensionless applied current density (DAC) with respect to the chemical oxygen demand (COD), considering the external source of COD brought by the added tylosin. The DAC is defined in Eq. (VII.72) [55-57]:

$$DAC = \frac{j_{\text{app}}}{j_{\text{lim,0}}} = \frac{j_{\text{app}}}{4Fk_{\text{m}}COD_{\text{tot}}}$$
(VII.72)

where  $j_{\text{lim},0}$  in A m<sup>-2</sup>,  $k_{\text{m}}$  equaled to  $2.33 \times 10^{-5}$  m s<sup>-1</sup> [58] and  $COD_{\text{tot}}$  is the sum of COD of wastewater (20.0 mg-O<sub>2</sub> L<sup>-1</sup>) and that of tylosin (8.8 mg-O<sub>2</sub> L<sup>-1</sup>) expressed in mol m<sup>-3</sup>.

Thus,  $j_{\text{lim},0}$  of the investigated effluent equaled to 0.8 mA cm<sup>-2</sup> (or 8 A m<sup>-2</sup>). Hence, the *DAC* of 1, 5 and 20 corresponded to the  $j_{\text{app}}$  of 0.8, 4 and 16 mA cm<sup>-2</sup> respectively, and they have been selected for this work.

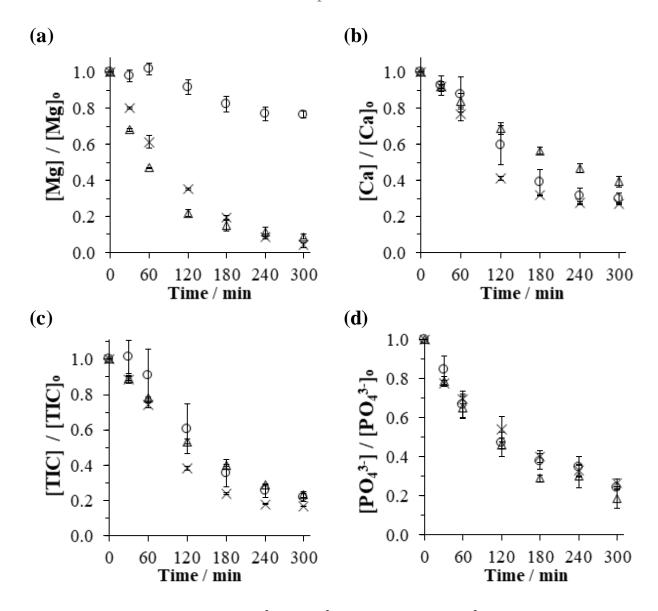
Figure VII.10 depicts the evolution of inorganic species during the electro-oxidation of real wastewater at the three investigated DAC. From Fig. VII.10(a), it was observed that the least

Mg(OH)<sub>2</sub> precipitation occurred at the lowest applied current (23.7  $\pm$  2.1%) and it increased to 95.8  $\pm$  2.9% at 4 mA cm<sup>-2</sup>. Using 16 mA cm<sup>-2</sup>, the kinetics of Mg(OH)<sub>2</sub> precipitation was higher until 4 h of electrolysis before the Mg settled at slightly lower overall precipitation (81.9  $\pm$  2.7%). Looking at the Ca precipitation in Fig. VII.10(b), Ca precipitated by about  $70.0 \pm 6.2\%$  using  $j_{app}$  of 0.8 mA cm<sup>-</sup>  $^{2}$ , increased to 73.2 + 1.1% using 4 mA cm<sup>-2</sup> and it dropped down to 60.6 + 2.8% using 16 mA cm<sup>-2</sup> <sup>2</sup>. Reproducible trend was found between the occurrence of Mg and Ca precipitates in the RW in function of DAC. They increased with increasing  $j_{app}$ . It could be due to more production of OH on cathode surface via the reduction reactions of dissolved  $O_2$  and water [5]. Nevertheless, when the  $j_{app}$ was too high, the overpotential on cathode became largely superior to the standard potential of dissolved O<sub>2</sub> reduction, and consequently, the H<sub>2</sub>O reduction reaction was dominant. The latter took place with concomitant HER; the higher the applied current, the higher the HER intensity. As a result, more H<sub>2</sub> bubbles evolved on cathode surface. On one hand, less active surface was available for: (1) the production of OH<sup>-</sup> to further react with Mg<sup>2+</sup> forming Mg(OH)<sub>2</sub> and (2): the production of OH<sup>-</sup> to shift HCO<sub>3</sub><sup>-</sup> to CO<sub>3</sub><sup>2</sup>- for subsequent reaction with Ca<sup>2+</sup> to produce CaCO<sub>3</sub>. Both resulted in lesser amount of Mg and Ca precipitations. On the other hand, vigorous concomitant H<sub>2</sub> evolution could also provoke the detachment of electro-precipitate already formed on cathode surface [5, 59, 60]. These reasoning explained why lesser electro-precipitate was formed at the highest  $j_{app}$ . Comparable trend can be reckoned from the evolution of TIC in Fig. VII.10(c) owing to its precipitation reaction with Ca<sup>2+</sup>. Nevertheless, it is worth mentioning that at high applied current density, carbonates could play the role of OH scavenger under the form of  $CO_3^{\bullet-}$  [61]. It would cause less carbonates present under the form of CO<sub>3</sub><sup>2-</sup> at the proximity of cathode to precipitate with Ca<sup>2+</sup>. This might also be the reason why less CaCO<sub>3</sub> was noticed at 16 mA cm<sup>-2</sup>. Next, Fig. VII.10(d) plots the evolution of PO<sub>4</sub><sup>3</sup>concentration during the electro-oxidation at all three applied currents. No significant difference was noted at the three investigated applied current. After 5 h of experiments,  $76.4 \pm 1.7\%$ ,  $73.9 \pm 4.2\%$ and 83.0  $\pm$  9.6% of PO<sub>4</sub><sup>3</sup>- precipitated in the increasing order of  $j_{app}$ . Lei et al. have shown that better Ca<sub>3</sub>(PO<sub>4</sub>)<sub>2</sub> could be recovered by applying lower current, owing to lesser degree of precipitation of CO<sub>3</sub><sup>2-</sup> with Ca<sup>2+</sup> [6, 8]. Thus, more free Ca<sup>2+</sup> would be able to react with PO<sub>4</sub><sup>3-</sup> [6, 8]. This deduction was not seen in these results due to perhaps low initial phosphate content to make significant difference.

The evolution of Cl species in the electrolytic system is plotted in **Figs. VII.11(a)**, **VII.11(c)** and **VII.11(e)**. At 0.8 mA cm<sup>-2</sup>, only 1.1  $\pm$  0.1% of the initially present Cl<sup>-</sup> was converted to principally ClO<sub>3</sub><sup>-</sup> (6.2  $\pm$  2.3 mg L<sup>-1</sup>; 2.6% of total Cl) after 5 h of electrolysis, with only 0.54  $\pm$  0.22 mg-Cl<sub>2</sub> L<sup>-1</sup> intermediary HClO/ClO<sup>-</sup> produced in the bulk (< 0.3% of total Cl). Low concentration of ClO<sub>4</sub><sup>-</sup> was present at this  $j_{app}$ ; 0.68  $\pm$  0.48 mg L<sup>-1</sup>, i.e. 0.24% of total Cl. Using higher applied current (4 mA cm<sup>-1</sup>)

<sup>2</sup>), 85.0 ± 2.6% of Cl<sup>-</sup> was oxidized leading to the production of 21.6 mg-Cl<sub>2</sub> L<sup>-1</sup> of HClO/ClO<sup>-</sup> after 2 h of electrolysis (i.e., 10.1% of total Cl). The reactions of Cl<sup>-</sup> and HClO/ClO<sup>-</sup> (Eqs. (VII.37)-(VII.38) and Eq. (VII.42)) would then lead to continuous production of  $ClO_3^-$  (188.0  $\pm$  2.5 mg L<sup>-1</sup> after 5 h of electrolysis), which was the main chlorinated byproduct using 4 mA cm<sup>-2</sup>. ClO<sub>3</sub><sup>-</sup> was also the main precursor to the production of  $ClO_4^-$  (Eqs. (VII.45) and (VII.68)). 6.1  $\pm$  1.1 mg L<sup>-1</sup> of  $ClO_4^$ was produced at the end of treatment (2.1% of total Cl) using 4 mA cm<sup>-2</sup>. At the highest investigated  $j_{\rm app}$  (16 mA cm<sup>-2</sup>), the highest decline in Cl<sup>-</sup> concentration was noted compared to the lower  $j_{\rm app}$ . It diminished completely after 3 h of electro-oxidation. The production of free chlorine only took place at the first half hour of the process, accumulating up to only  $5.55 \pm 0.46$  mg-Cl<sub>2</sub> L<sup>-1</sup> (i.e., 2.8% of total Cl) before being consumed rapidly. Up to 1 h of electrolysis, Cl<sup>-</sup> and free chlorine were rapidly converted to  $ClO_3^-$ , up to  $168 \pm 2$  mg L<sup>-1</sup> in the bulk (representing 67.4% of total Cl).  $ClO_4^$ progressively started to accumulate after 0.5 h of electro-oxidation process. ClO<sub>3</sub><sup>-</sup> as the intermediate started to decrease after 1 h of electrolysis and the accumulation of  $ClO_4$  went up to  $223 \pm 1$  mg L<sup>-1</sup> (71.3% of total chlorine) after 5 h of treatment. This time, ClO<sub>4</sub> was the dominant chlorinated byproduct of electro-oxidation treatment when 16 mA cm<sup>-2</sup> was adopted. Less than 12.6% of Clvolatile was produced using 0.8 mA cm<sup>-2</sup>, about 16.0% using 4 mA cm<sup>-2</sup> and 43% when using 16 mA cm<sup>-2</sup>. ClO<sub>2</sub> was always below the value of limit of quantification owing to its rapid kinetics of reaction as previously discussed in Section VII.3.2. In the work of Bergmann et al. [16], a synthetic solution initially containing 50 mg L<sup>-1</sup> of Cl<sup>-</sup> was electrolyzed using BDD rotating disk anode of 35 mm diameter (~10 cm<sup>2</sup> surface area, IrO<sub>2</sub> as cathode), at 20 mA cm<sup>-2</sup>. In this condition, 9.5 mg L<sup>-1</sup> of active chlorine was produced after 10 min of electrolysis before its concentration dropped drastically to yield 30 mg L<sup>-1</sup> of ClO<sub>3</sub><sup>-</sup>. Lower ClO<sub>3</sub><sup>-</sup> was expected due to lower initial Cl<sup>-</sup> content. In similar work, when drinking water (initial Cl<sup>-</sup> content of 41 ppm) was electrolysed using similar electrolytic setup, 120 mg L<sup>-1</sup> ClO<sub>4</sub> was reported. In other work from Lacasa et al., an electrolyte initially containing 300 mg L<sup>-1</sup> Cl<sup>-</sup> was oxidized at 30 mA cm<sup>-2</sup> in a cell with 78 cm<sup>2</sup> BDD (stainless steel as cathode) [19]. Very quick conversion of free chlorine to ClO<sub>3</sub><sup>-</sup> occurred after 30 min of electrolysis producing 351 mg-ClO<sub>3</sub><sup>-</sup> L<sup>-1</sup>. Upon reaching this maximum concentration, ClO<sub>3</sub><sup>-</sup> was converted to ClO<sub>4</sub> and the latter reached its maximum concentration of 842 mg-ClO<sub>4</sub> L<sup>-1</sup> after 2 h of electrolysis. In another work from Scialdone et al. [62], in a system where chloroacetic acid (473 mg L<sup>-1</sup>) was treated electrochemically in a cylindrical undivided cell using BDD anode and stainless steel cathode (both possessing 3 cm<sup>2</sup> surface area), ClO<sub>3</sub><sup>-</sup> and ClO<sub>4</sub><sup>-</sup> were dosed at different applied current (20, 40 and 100 mA cm<sup>-2</sup>). Approximately 232, 116 and 62 mg-ClO<sub>3</sub>- L<sup>-1</sup> were produced in the ascending order of the investigated  $j_{app}$ . In the meantime, about 10 mg-ClO<sub>4</sub>- L<sup>-1</sup> was produced using 20 mA cm<sup>-</sup> <sup>2</sup> and 472 mg-ClO<sub>4</sub>- L<sup>-1</sup> was formed at both 40 and 100 mA cm<sup>-2</sup>. Separately, Herraiz-Carboné et al. reported a production of up to 49.6 mg-Cl<sub>2</sub> L<sup>-1</sup> of HClO/ClO<sup>-</sup> at 2.5 mA cm<sup>-2</sup>, while treating synthetic urine wastewater [41]. No presence of ClO<sub>3</sub><sup>-</sup> and ClO<sub>4</sub><sup>-</sup> at all was detected under their experimenting conditions (up to 5 mA cm<sup>-2</sup>, BDD and stainless steel as anode and cathode, respectively). Upon comparison with literatures, suitable analogy could not be done due to dissimilarities in reactor design, applied operating conditions as well as initial chloride content in the electrolyte. Nonetheless, general conclusion of the evolution of chlorinated species could be drawn: (1) the reactivity of HClO/ClO<sup>-</sup> as intermediate product was rapid and it was quicker with increased current density and (2) the concentration of intermediate ClO<sub>3</sub><sup>-</sup> as well as the subsequent ClO<sub>4</sub><sup>-</sup> product were highly dependent on the initial Cl<sup>-</sup> content as well as the current density adopted.

Figures VII.11(b), VII.11(d) and VII.11(f) depict the distribution percentages of N-species as a function of electro-oxidation treatment time at 0.8, 4 and 16 mA cm<sup>-2</sup>. At the three investigated applied currents, the concentration of NO<sub>3</sub><sup>-</sup> decreased slightly at the beginning of the process leading to the formation of NH<sub>4</sub><sup>+</sup> and N-volatile (Eqs. (VII.9)-(VII.10)) before reaching equilibrium in the bulk. No accumulation of NO<sub>2</sub> was detected throughout the process, due to its low concentration and quick conversion to NO<sub>3</sub><sup>-</sup> (Eqs. (VII.12) and (VII.16)) and NH<sub>4</sub><sup>+</sup> (Eq. (VII.19)). At the lowest current density (0.8 mA cm<sup>-2</sup>), NH<sub>4</sub><sup>+</sup> was only partially removed in the electrolytic system, which was not the case at higher current density. It decreased by about 29.7% at the end of the electro-oxidation process. This finding can be attributed to lower conversion of Cl<sup>-</sup> to HClO/ClO<sup>-</sup> (Fig. VII.11(a)). Using 4 mA cm<sup>-2</sup>, NH<sub>4</sub><sup>+</sup> reacted rapidly with HClO/ClO<sup>-</sup> to form chloramines. At the highest applied current density tested (16 mA cm<sup>-2</sup>), NH<sub>4</sub><sup>+</sup> was quickly consumed. However, starting from 1 h of treatment, NH<sub>4</sub><sup>+</sup> increased again. This particular observation was attributed to the depletion of HClO/ClO<sup>-</sup> in the bulk (**Fig. VII.11(e)**), thus there was less reactivity with NH<sub>4</sub><sup>+</sup> to form chloramines (Eqs. (VII.26)-(VII.28)). Furthermore, the higher accumulation of ClO<sub>4</sub> in the bulk did not help to oxidize NH<sub>4</sub><sup>+</sup>. Its reactivity with NH<sub>4</sub><sup>+</sup> was lower than HClO and Cl<sub>2</sub> due to its high activation energy [19]. In addition, slightly higher production rate of NH<sub>4</sub><sup>+</sup> than its reduction to NO<sub>3</sub><sup>-</sup> and/or N<sub>2</sub> in the real effluent led to its accumulation in the electrolytic system. Moreover, when the production of Nvolatile was compared at different  $j_{app}$  (Figs. VII.11(b), VII.11(d) and VII.11(f)), it was found that N-volatile production increased with increasing applied current. 10.8, 31.1 and 40.3% of N-volatile were generated in the increasing order of the investigated  $j_{app}$ . N-combined (i.e., inorganic chloramines and organochloride compounds) was the major form of N-bearing byproducts after the electro-oxidation treatment of the tylosin-spiked wastewater effluent at all investigated  $j_{app}$ . The degradation and mineralization degree of the targeted model organic micropollutant is discussed in Section VII.3.5.2.



**Fig. VII.10**. Concentrations of (a)  $Mg^{2+}$ , (b)  $Ca^{2+}$ , (c) TIC and (d)  $PO_4^{3-}$  during the electrolysis of real effluent spiked with 10 mg  $L^{-1}$  tylosin at (O): 0.8, (×): 4 and ( $\Delta$ ): 16 mA cm<sup>-2</sup>. Anode: BDD and cathode: stainless steel.

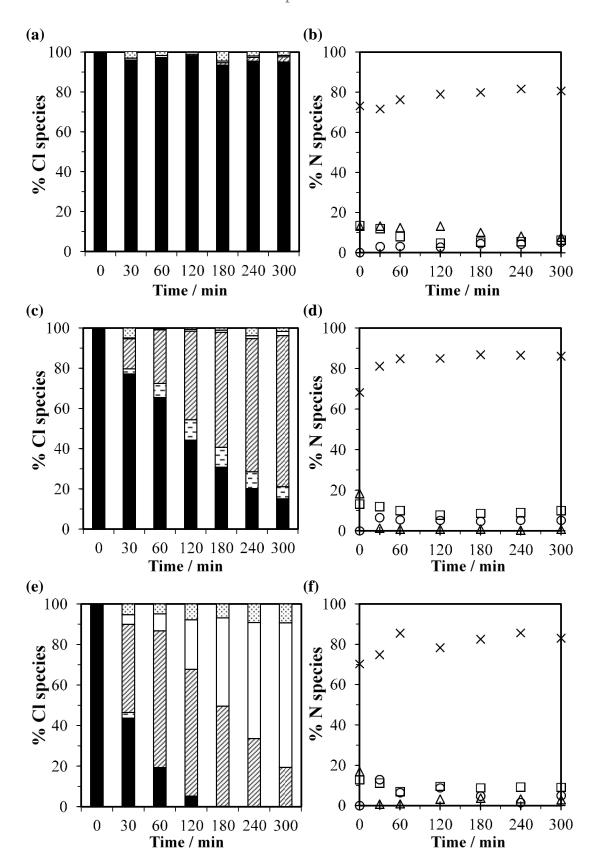


Fig. VII.11. Percentage of distributions of Cl-species during the electro-oxidation of real effluent spiked with 10 mg L<sup>-1</sup> tylosin at (a) 0.8, (c) 4 and (e) 16 mA cm<sup>-2</sup> as well as the distribution of N-species at (b) 0.8, (d) 4 and (f) 16 mA cm<sup>-2</sup>. (□): Cl<sup>-</sup>, (□): HClO/ClO<sup>-</sup>, (□): ClO<sub>2</sub><sup>-</sup>, (□): ClO<sub>3</sub><sup>-</sup>, (□): ClO<sub>4</sub><sup>-</sup> and (□): Cl-volatile, while (△): NH<sub>4</sub><sup>+</sup>, (+): NO<sub>2</sub><sup>-</sup>, (□): NO<sub>3</sub><sup>-</sup>, (O): N-volatile and (×): N-combined. Anode: BDD and cathode: stainless steel.

# VII.3.5.2. Kinetics and modeling of tylosin degradation as target micropollutant in reclaimed wastewater at different applied current densities

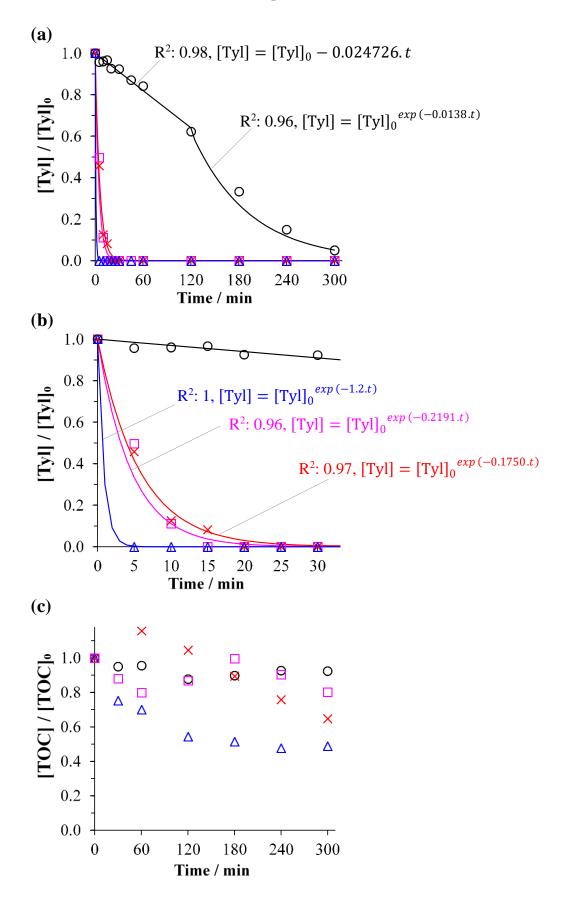
The plot of tylosin degradation as a function of time at the three  $j_{app}$  is given in Fig. VII.12. It can be observed that when the applied current corresponded to DAC=1 with respect to  $j_{\text{lim},0}$  ( $j_{\text{app}} = 0.8 \text{ mA}$ cm<sup>-2</sup>), the apparent degradation of tylosin decreased quasi-linearly over time up to 2 h of electrolysis. It was then followed by slower kinetics during the last three hours of electro-oxidation to reach 95% of abatement. This linear trend during the first 2 h of electro-oxidation was expected due to the applied  $j_{\text{app}}$ , which was lower or equal to the  $j_{\text{lim},0}$  of the initial COD of the tylosin. As a consequence, the oxidation of tylosin was limited by charge transfer [56, 57, 63]. During this period, tylosin degradation followed a zero-order kinetic rate with a kinetic constant of 0.0247 mg L<sup>-1</sup> min<sup>-1</sup> (R<sup>2</sup> = 0.98) (Fig. VII.12(a)). When the tylosin concentration decreased with treatment time, its limiting current density at time  $t(j_{\text{lim.t}})$  also decreased. Once the  $j_{\text{app}}$  equaled to or was higher than the  $j_{\text{lim.t}}$ , tylosin degradation became mass transfer limited. As a result, its apparent kinetic degradation was no longer zero-order rate but followed pseudo-first order rate law [56, 57, 63]. From 2 h onwards (Fig. VII.12(a)), its decrease gave out first-order rate constant of 0.0138 min<sup>-1</sup> ( $R^2 = 0.96$ ). The deviation of modeled tylosin degradation curve at 0.8 mA cm<sup>-2</sup> was compared with the experimental curve using RMSE (Eq. (III.3)). A deviation of 0.024 was evaluated which means that the theoretical model was conformed to the experimental data.

In contrast, using higher  $j_{app}$  of 4 mA cm<sup>-2</sup> (DAC = 5), total degradation of tylosin parent molecule in RW was observed after 20 min of treatment (Fig. VII.12(a)-(b)). The degradation followed pseudofirst order kinetic rate with a rate constant of 0.17 min<sup>-1</sup>. Tylosin abatement in simulated effluent (spiked with tylosin at 10 mg L<sup>-1</sup>) at comparable current density was also investigated (Fig. VII.12). Slightly faster kinetics of elimination was noticed in the simulated matrix. Tylosin molecule was totally removed from the effluent after 15 min of electro-oxidation process. Consequently, the evolution of tylosin elimination could be fitted with marginally higher kinetic rate constant (i.e., 0.2191 min<sup>-1</sup>) compared to the matrix of RW. Previously in Section VII.3.4, it has been demonstrated that more electro-precipitation took place in real wastewater effluent due to the more complex organic matrix in real waste effluent. Nevertheless, it has also been shown in Chapter 6 that during 5-h of electro-oxidation, mineral scaling did not totally passivate the cathode surface. R<sub>CT</sub> on electrode surface remained constant throughout the treatment. It suggests that the precipitation layer formed did not yet give major impact towards electronic exchanges for direct and mediated target pollutant degradation or organics mineralization in general. No fouling or deposition was observed on the anode. In conjunction with that, the slightly slower tylosin degradation observed in Fig. VII.12(a)-VII.12(b) was rather due to the presence of recalcitrant dissolved organic matter in the real

wastewater matrix. Its presence could compete with the degradation of tylosin molecule which led to slower removal of the target pollutant [41, 53]. The evolution of TOC during the electrolysis illustrated in Fig. VII.12(c) could give some insight towards the hypothesis. Under the polarization of 4 mA cm<sup>-2</sup>, it was observed that more mineralization occurred in the real (35.1%) compared to the simulated (19.7%) effluents. The increase of TOC at the beginning of electro-oxidation was perhaps due to the breakdown of large tylosin parent compound (MW: 1066 g mol<sup>-1</sup>). This resulted in an increase of dissolved organic content in the treated effluent. Here, it confirmed the competition between tylosin and dissolved organics since more mineralization occurred but slightly slower tylosin was removed. In both cases, the modeled curves of tylosin removal fitted well the experimental counterparts. RMSE were calculated at 0.021 and 0.048 in the real and simulated wastewater respectively. When the highest tested current density was applied (16 mA cm<sup>-2</sup>), very quick tylosin degradation was observed. Total tylosin removal was already spotted during the first 5 min of treatment (Fig. VII.12(a)-(b)) with a kinetic rate constant of 1.2 min<sup>-1</sup>.

Quick degradation of tylosin parent molecule demonstrated in this work corroborated the efficient treatment of tylosin by various advanced oxidation technologies documented in literature [47, 49, 64-67]. The only EAOP process to degrade tylosin was by EF reported by Ferrag-Siagh et al. [49]. 600 mL of synthetic solution containing 100 mg L<sup>-1</sup> of tylosin and 300 mA applied current were used during the process. Complete tylosin degradation was noticed after 15 min of treatment. The tylosin degradation followed first-order kinetic law with 0.3083 min<sup>-1</sup> rate constant. The kinetic rate constant was comparable with the one obtained in this work (i.e., 0.1750 to 0.2191 min<sup>-1</sup> using 200 mA applied current). Slightly higher value reported in their work was due to higher applied current and could also be due to better performance of EF against EO as suggested by several authors [68, 69].

In summary, it has been shown that after 5 h of electro-oxidation treatment, the quantity of electro-precipitation depended on the  $j_{app}$ . Nonetheless, the mineral scaling formed on cathode surface did not result in significant reduction in electronic transfer on electrode surface (after 5-h operation). No fouling or deposition was observed on anode. While treating real reclaimed wastewater, higher overall organic abatement was achieved but with slower target pollutant removal. Slower kinetics of the latter was due to the competition between persistent dissolved organics contained in the real wastewater matrix. Either way, the removal of tylosin using EAOP was efficient. Its removal could be estimated using first-order kinetic law but the kinetic constant might vary between EAOPs as function of applied operating conditions.



**Fig. VII.12**. Kinetics and modeling of (a) tylosin degradation, (b) zoom into the early kinetics of tylosin degradation and (c) mineralization during the electro-oxidation of RW containing 10 mg L<sup>-1</sup> tylosin. (O, —): 0.8, (×, —): 4 and ( $\triangle$ , —): 16 mA cm<sup>-2</sup>. ( $\square$ , —): simulated wastewater (SW) spiked with 10 mg L<sup>-1</sup> tylosin (4 mA cm<sup>-2</sup>). Anode: BDD and cathode: stainless steel.

#### VII.4. Conclusions

A bottom-up approach has been adopted to investigate thoroughly the influence of complex matrices of electrolyte on the formation of electro-precipitate, the degradation of target pollutant and overall organics mineralization inside microfluidic reactor. The main new insights are given hereinafter:

- Only Mg(OH)<sub>2</sub>, CaCO<sub>3</sub> and Ca<sub>3</sub>(PO<sub>4</sub>)<sub>2</sub> were formed during the electro-oxidation process while the formation of struvite, MgCO<sub>3</sub>, CaMg(CO<sub>3</sub>)<sub>2</sub> and Mg<sub>3</sub>(PO<sub>4</sub>)<sub>2</sub> were not favoured. Similar deposition elements were found between submillimetric (this work) and centimetric (literatures) electrochemical reactor setups. Therefore, the anodic contribution in microfluidic configuration only affected the kinetics of electro-precipitation.
- The addition of multi-ions and organic matter onto the SW brought insignificant effect on the formation of electro-precipitate during the electrochemical treatment. High cathodic local pH dissipated the inhibiting capacity of organics towards the Mg(OH)<sub>2</sub>, CaCO<sub>3</sub> and Ca<sub>3</sub>(PO<sub>4</sub>)<sub>2</sub> deposition.
- More mineral scaling occurred in RW compared to SW. It could be due to either (1) the presence of more complex dissolved organics in the wastewater that could most probably co-precipitate with Mg<sup>2+</sup> and Ca<sup>2+</sup> or (2) the high sulfate content in the simulated effluent that would inhibit on Mg and Ca precipitation.
- After 5 h of treatment, electro-deposition formed on cathode surface did not decrease the operation
  efficiency. Higher organic mineralization but slightly slower target pollutant degradation was
  noticed. The contrasting trend was attributed to the competition between persistent micropollutants
  with the target pharmaceuticals in the treated effluent.
- By applying the highest  $j_{app}$  (16 mA cm<sup>-2</sup>), 51% of mineralization, 100% tylosin degradation and 45 mg of electro-precipitate were obtained after 5 h of electro-oxidation. Higher organic mineralization and lower mineral deposition were obtained in comparison with lower  $j_{app}$ .
- The highest energy consumption was procured using 16 mA cm<sup>-2</sup> (41.2 kWh m<sup>-3</sup> vs. 8.33 kWh (4 mA cm<sup>-2</sup>) and 1.29 kWh m<sup>-3</sup> (0.8 mA cm<sup>-2</sup>)). Moreover, the highest quantity of undesired byproducts (ClO<sub>3</sub><sup>-</sup> + ClO<sub>4</sub><sup>-</sup>) was also obtained under this condition (151 mg-Cl<sub>2</sub> L<sup>-1</sup> vs. 82 (4 mA cm<sup>-2</sup>) and 2.9 (0.8 mA cm<sup>-2</sup>) mg-Cl<sub>2</sub> L<sup>-1</sup>).

Upon these results, a  $j_{app}$  in the region of 1 to 2 mA cm<sup>-2</sup> might be good compromise between organics abatement, scaling deposition and the yield of nuisance byproducts. The treatment duration ought to be longer to efficiently mineralize organic content. It would not be problematic to the submilimetric reactor judging from its very low electrical power consumption (e.g. 1.29 kWh m<sup>-3</sup> during 5 h electro-oxidation at 0.8 mA cm<sup>-2</sup>). Using the suggested  $j_{app}$  (1 to 2 mA cm<sup>-2</sup>), the decrease in electro-oxidation

treatment efficiency by mineral scaling for a long-period operation should also be re-investigated more practically under continuous operating mode.

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## VIII. Influence of cathode materials towards the formation of electroprecipitate

#### VIII.1. Introduction

Carbonaceous materials such as carbon felt, graphite felt and carbon sponge are well-known to inherit high porosity features that can overcome mass transfer limitation owing to their great specific surface [1]. They have also been the subject of research due to their high electrocatalytic properties, while offering low operating cost and environmentally-friendly properties compared to noble metals [2]. This can benefit wide range of prospective applications [1, 3, 4], including in the water treatment domain [5, 6].

In this chapter, the influence of the properties of carbonaceous materials on the occurrence of cathodic mineral scaling during the advanced electro-oxidation treatment of simulated and reclaimed wastewater was investigated for the first time. The cathode materials were firstly characterized using an electrochemical method before being subjected to electrolysis at different applied current densities. The role of electrode porosity was particularly explored and the amount of mineral electroprecipitation produced was systematically quantified.

Besides, another distinct characteristic of carbonaceous materials is that they possess lower overpotential for water reduction [7]. Consequently, wider potential window under cathodic range is achieved, prior to HER. Motivated by such additional interesting property of carbon material, the role of overpotential of  $H_2$  evolution on the kinetics of electro-precipitation was further examined. The synergistic effect of HER overpotential coupled with the electrochemical configuration were discussed. The comparison between the occurrence of mineral scaling on different cathode materials inside submillimetric and millimetric  $d_{\text{elec}}$  was evaluated. Moreover, the new insights given by mechanistic understanding towards the electrochemical and chemical processes taking place at the electrode/electrolyte interface was substantiated with the *in-situ* electrochemical impedance analysis.

### VIII.2. Experimental section

Stainless steel, graphite or carbon paper was used as cathode to investigate the influence of electrode material on the formation of electro-precipitate. CV was performed in a three-electrode configuration to estimate the electroactive surface of the different materials under study. 500 mL of solution containing 50 mM of K<sub>4</sub>Fe(CN)<sub>6</sub> in 0.50 M of Na<sub>2</sub>CO<sub>3</sub> was used [7, 8]. The cathode material to be characterized was used as WE, BDD was the CE and Ag-AgCl was the RE. The potential was swept between 0.6 and -0.8 V/Ag-AgCl with a scan rate of 10 mV s<sup>-1</sup>. Interelectrode gap of 500 µm was

implemented during this characterization experiments. The electrolyte was circulated at 100~mL min  $^{-1}$ 

The electro-oxidation experiments were carried out using 500 mL solution containing 150 mg L<sup>-1</sup>  $Ca^{2+}$ , 5 mg L<sup>-1</sup>  $Mg^{2+}$  and 60 mg-C L<sup>-1</sup> of TIC. It corresponds to the BE such used in **Chapters 4**, 6 and 7. Galvanostatic electrolysis using two-electrode configuration was used to run the experiments. Two values of applied current were investigated: 20 (0.4 mA cm<sup>-2</sup>) and 200 mA (4 mA cm<sup>-2</sup>). BDD served as anode while stainless steel, graphite and carbon paper were used as cathode. The solution was recirculated at a flow rate of 100 mL min<sup>-1</sup>. During the study of the role of interelectrode gap, three  $d_{\rm elec}$  were investigated: 100 and 500  $\mu$ m, representing submillimetric configurations, and 3000  $\mu$ m representing the millimetric scale.

To characterize the electrochemical behavior of BE on the investigated cathodes during electrolysis, LSV was performed via a potentiostat under a three-electrode setup. A potential sweep between 0.5 and -1.8 V/Ag-AgCl with 10 mV s<sup>-1</sup> scan rate was conducted. Stainless steel, graphite and carbon paper were used as the WE, BDD as CE and Ag-AgCl as RE. EIS was conducted in galvanostatic mode covering the frequency ranges between 100 kHz to 100 mHz. The perturbation amplitude ( $\Delta I_{AC}$ ) varied between 5 and 10 mA RMS according to EIS spectra with 10 ppd. The EIS analysis was programmed at regular interval during galvanostatic electrolysis at an applied direct current of 200 mA (i.e. 4 mA cm<sup>-2</sup>). The EIS parameters were extracted by means of EEC model using ZSimpwin® commercial software.

Detailed analytical methods have been described in **Chapter 3**. Mg and Ca concentrations were quantified using ICP-OES. The evolution of TIC concentration was followed using TOC-TN analyzer. The morphology of electro-precipitate on carbon paper cathode surface was performed using SEM-EDX. During the electro-oxidation experiments of reclaimed wastewater (**Section VIII.3.1.3**), multi-ions were present. The evolution of anion concentrations (NO<sub>3</sub><sup>-</sup>, NO<sub>2</sub><sup>-</sup>, SO<sub>4</sub><sup>2</sup>-, Cl<sup>-</sup>, ClO<sub>2</sub><sup>-</sup>, ClO<sub>3</sub><sup>-</sup> and ClO<sub>4</sub><sup>-</sup>) were measured using ionic chromatography. NH<sub>4</sub><sup>+</sup> was measured with HACH spectrophotometer, PO<sub>4</sub><sup>3</sup>- was quantified using HACH PhosVer 3 method and free chlorine was dosed using HACH DPD method. TOC and TN complemented TIC analysis using TOC-TN analyzer (Shimadzu). Tylosin concentration was monitored using HPLC-UV (wavelength 290 nm).

#### VIII.3. Results and discussion

## VIII.3.1. Influence of the porosity of cathode material on the formation of electroprecipitate

In this section, the influence of cathode porosity on the kinetics of electro-precipitate formation was thoroughly evaluated. Electrodes of different porosities, namely stainless steel, graphite and carbon paper were used as case study. These electrodes were firstly characterized by means of electrochemical method in **Section VIII.3.1.1**, before they were used as cathode and subjected to electrolysis at different applied currents (20 and 200 mA) in **Section VIII.3.1.2**. Finally, **Section VIII.3.1.3** was dedicated to a case study where the performance of porous cathode (carbon paper) as well as non-porous cathode (stainless steel) was compared regarding the electro-oxidation efficiency of reclaimed wastewater. The formation of electro-precipitation together with the fate of inorganic and organic species in the treated media were quantified.

# VIII.3.1.1. Electroactivity of stainless steel, graphite and carbon paper characterized by electrochemical method

Redox reactions of Fe(CN<sub>6</sub>)<sup>3-</sup> and Fe(CN<sub>6</sub>)<sup>4-</sup> were used to estimate the electroactivity of the three investigated cathode surfaces. **Figure VIII.1** illustrates the voltammograms of stainless steel, graphite and carbon paper when they were subjected to CV. The peak current ( $I_P$  in A) is correlated to the electroactive surface area of an electrode by Randles-Sevcik equation (**Eq. (VIII.1)**) [7-9]:

$$I_{\rm P} = 2.69 \times 10^5 \times AD_{\rm L}^{1/2} n_{\rm e}^{3/2} \gamma^{1/2} C_{\rm SOL}$$
 (VIII.1)

where A is the electroactive surface area (in cm<sup>2</sup>),  $D_L$  is the diffusion coefficient of Fe(CN<sub>6</sub>)<sup>3</sup>-/Fe(CN<sub>6</sub>)<sup>4-</sup> in solution (6.48 × 10<sup>-6</sup> cm<sup>2</sup> s<sup>-1</sup> [10-12]),  $n_e$  equals to 1,  $\gamma$  is the scan rate (0.01 V s<sup>-1</sup>) and  $C_{SOL}$  is 5 × 10<sup>-5</sup> mol cm<sup>-3</sup>.

From the values of  $I_P$  measured for the three cathode materials (**Fig. VIII.1**), their corresponding values of electroactive surface areas were determined. It was estimated that the stainless steel (**Fig. VIII.1(a)**), graphite (**Fig. VIII.1(b)**) and carbon paper (**Fig. VIII.1(c)**) possessed 80, 85 and 175 cm<sup>2</sup> of surface area, respectively. This trend is coherent with the increase of electrode porosity from stainless steel to graphite ( $\sim$ 15%, Final Advanced Materials, Didenham, France) and carbon paper ( $\sim$ 89%, Ion Power, München, Germany). There was only 6% difference between graphite and stainless steel despite the former being a carbonaceous material. This slightly higher electroactive surface of graphite was due to the fact that both stainless steel and graphite used in this work were solid plate electrodes. Porous properties of graphite gave slight advantage compared to the stainless

steel plate. Significant increase of electroactive surface area was provided by the carbon paper as expected. It was attributed to its high porosity which increased the number of electroactive sites at the cathode interface.

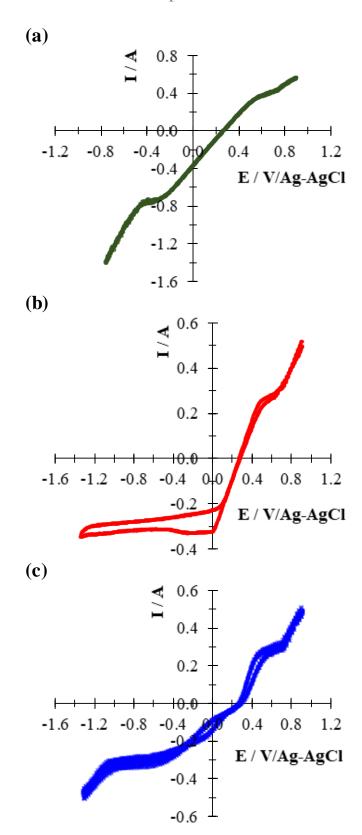
The LSV of the three investigated cathode materials in BE (150 mg L<sup>-1</sup> Ca<sup>2+</sup>, 5 mg L<sup>-1</sup> Mg<sup>2+</sup> and 60 mg-C L<sup>-1</sup> of TIC) used to study the formation of electro-precipitation are given in **Fig. VIII.2**. From the LSV curves, the exchanged current ( $I_0$ ) on stainless steel, graphite and carbon paper were estimated to be 1.2, 7.9 and 15.8 mA, respectively.  $I_0$  indicated the easiness of electrons transfer that could occur at the electrode/electrolyte interface. From the values of  $I_0$ , charge transfer resistance ( $R_{CT}$ ) could hence be estimated using **Eq. (VIII.2) [13, 14]**:

$$R_{\rm CT} = \frac{RT}{n_{\rm e}FI_0} \tag{VIII.2}$$

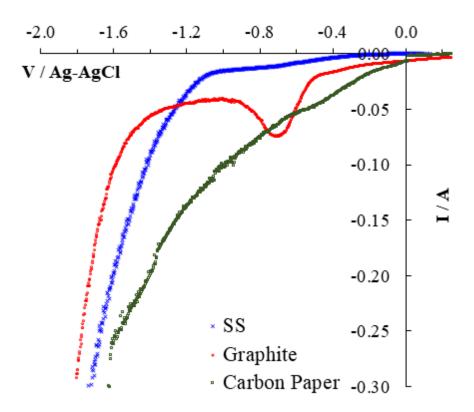
where  $R_{\text{CT}}$  is in  $\Omega$ , R is the gas constant (8.314), T is the temperature (in K) and  $n_{\text{e}}$  is the number of exchanged electrons for water reduction. Here,  $n_{\text{e}}$  equals to 2, due to the fact that water reduction by 2 electrons was the dominant faradaic reaction at the applied current investigated [15].

By intermediary of Eq. (VIII.2),  $R_{\rm CT}$  on stainless steel, graphite and carbon paper were estimated to be 10.2, 1.62 and 0.81  $\Omega$ , respectively. These numbers foretell that electronic transfer is the easiest on carbon paper, followed by graphite and finally on stainless steel cathode. It tallies with the respective order of the above-mentioned surface of materials. The values of  $R_{\rm CT}$  are compared with those measured with another technique (i.e., EIS) in Section VIII.3.2.2.

In the meantime, the role of cathode material on the kinetics of electro-precipitate during the electrolysis of solution containing 150 mg  $L^{-1}$   $Ca^{2+}$ , 5 mg  $L^{-1}$   $Mg^{2+}$  and 60 mg-C  $L^{-1}$  of TIC is scrutinized in the next section.



**Fig. VIII.1**. Voltammograms on (a) carbon paper, (b) graphite and (c) stainless steel for the determination of electroactive surface area. Electrolyte: 500 mL of 50 mM of  $K_4Fe(CN)_6$  in 0.50 M of  $Na_2CO_3$ ; potential sweep between 0.6 to -1.6 V/Ag-AgCl with a scan rate of 10 mV s<sup>-1</sup>; CE: BDD and RE: Ag-AgCl;  $d_{elec}$ : 500 μm.



**Fig. VIII.2**. LSV of stainless steel, graphite and carbon paper. 500 mL solution containing 150 mg  $L^{-1}$  Ca<sup>2+</sup>, 5 mg  $L^{-1}$  Mg<sup>2+</sup> and 60 mg-C  $L^{-1}$  of TIC. Potential sweeps between 0.5 to -1.8 V/Ag-AgCl with a scan rate of 10 mV s<sup>-1</sup>; CE: BDD and RE: Ag-AgCl;  $d_{elec}$ : 500 μm.

# VIII.3.1.2. Electro-precipitation on different cathode materials at various applied current densities

In this section, the formation of electro-precipitate occurring on stainless steel, graphite and carbon paper under the polarization of different applied currents was evaluated. **Figure VIII.3** illustrates some photographs of mineral electro-precipitate that were formed on the investigated cathodes when BE was polarized at 200 mA. The scaling appeared whitish on stainless steel (**Fig. VIII.3(a)**) and graphite (**Fig. VIII.3(b)**), whereas the white appearance was not seen on carbon paper (**Figs. VIII.3(c)-VIII.3(d)**). Perhaps, the electro-precipitation took place within the pores using carbon paper. With the occurrence of mineral scaling being proven, its quantitative analysis on different cathode materials and applied current was conducted. The evolution of  $Mg^{2+}$ ,  $Ca^{2+}$  and TIC concentrations during the 5 h of electrolysis at 20 and 200 mA are plotted in **Fig. VIII.4**. In **Fig. VIII.4**(a),  $Mg^{2+}$  concentration decreased insignificantly for all three investigated cathode materials when 20 mA was applied. There was  $5.2 \pm 0.1\%$  of decrease of  $Mg^{2+}$  concentration using stainless steel with 7% more decrease using graphite. Meanwhile, using carbon paper,  $Mg^{2+}$  concentration dropped by 2.1% after 3 h electrolysis before it re-dissolved into the solution. From these results, it could be deduced that little  $Mg(OH)_2$  precipitation occurred on all cathode materials investigated at low applied current. With respect to the evolution of  $Ca^{2+}$  concentration (20 mA) as shown in **Fig.** 

VIII.4(c), it decreased indifferently when stainless steel and graphite cathodes were used considering the standard deviations. It decreased to  $45.8 \pm 3.3\%$  and to  $47.8 \pm 1.0\%$  using stainless steel and graphite, respectively. Similar trend as the evolution of  $Mg^{2+}$  was observed in the case of carbon paper.  $Ca^{2+}$  concentration dropped by  $11.2 \pm 0.8\%$  after 3 h of electrolysis before dissolving back into solution to yield only 1.3% of Ca precipitation on carbon paper at the end of electrolysis. The evolution of TIC concentration is given in **Fig. VIII.4(e)**. Their evolutions followed those of  $Ca^{2+}$  plotted in **Fig. VIII.4(c)** with the exception of carbon paper. It was observed that all TIC concentrations depleted after 5 h of electrolysis:  $90 \pm 18\%$ ,  $94 \pm 14\%$  and  $94 \pm 7\%$  of carbonates loss in the setup using stainless steel, graphite and carbon paper, respectively. Using stainless steel and graphite, the decrease of carbonates in conjunction with  $Ca^{2+}$  concentration confirmed the  $CaCO_3$  electro-precipitation on both cathodes. Contrastingly, using carbon paper polarized at 20 mA, the TIC in the solution depleted but  $Ca^{2+}$  only precipitated by 1.3% at the end of electrolysis. Therefore,  $HCO_3^-/CO_3^{2-}$  might loss via an additional route, which is discussed later in this sub-section.

The evolution of Mg<sup>2+</sup>, Ca<sup>2+</sup> and TIC concentrations on stainless steel, graphite and carbon paper at higher applied current (200 mA) are shown in Figs. VIII.4(b), VIII.4(d) and VIII.4(f), respectively. From Fig. VIII.4(b), it was observed that similar degree of precipitation of Mg<sup>2+</sup> took place on the surface of stainless steel and graphite (91.0  $\pm$  1.8% and 89.1  $\pm$  2.2%, respectively). In contrast, lower  ${\rm Mg^{2+}}$  precipitation was observed on carbon paper (i.e.  $20.2\pm1.1\%$ ). With regard to the evolution of  $Ca^{2+}$  concentration at 200 mA (Fig. VIII.4(d)),  $Ca^{2+}$  precipitated by 43.2  $\pm$  5.6% and 46.7  $\pm$  1.2% after 5 h of electrolysis when stainless steel and graphite were used as cathode, respectively. Thus, 3.5% more CaCO<sub>3</sub> precipitated on graphite in comparison to stainless steel. When carbon paper was adopted as cathode,  $Ca^{2+}$  precipitated by  $48.2 \pm 1.3\%$  after just only 1 h of electrolysis. However, the precipitate re-dissolved into the bulk after 1 h of treatment and it ended with  $40.0 \pm 1.5\%$  of Ca<sup>2+</sup> precipitation, which was less than those found on stainless steel and graphite. Figure VIII.4(f) plots the evolution of TIC using the three cathode materials under the polarization of 200 mA. Comparable trends seen in the cases with 20 mA were observed. More precipitation of carbonates occurred using the graphite cathode setup (95.8  $\pm$  5.4%) versus the stainless steel (89.4  $\pm$  0.2%). This corroborated with the evolution of Ca<sup>2+</sup> concentration where slightly more Ca<sup>2+</sup> precipitated on graphite relatively to stainless steel. Regarding the TIC evolution under the carbon paper setup, as it was the case with  $Ca^{2+}$ , TIC depleted quickly down to 94.6  $\pm$  0.9% just after 1 h of electrolysis. TIC decreased significantly faster at higher applied current in comparison to the case where 20 mA was applied (i.e., 4 h to deplete the TIC by 94% (Fig. VIII.4(e))).

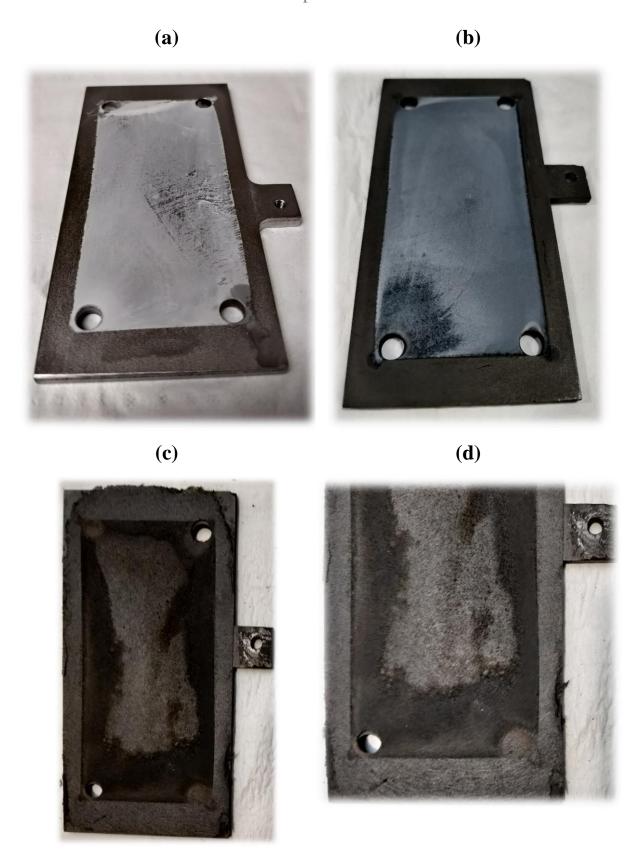
The peculiarity of the results obtained when carbon paper was adopted as cathode at both applied current could be justified by the trend of electrolytic solution pH over the course of electrolysis.

**Figure VIII.5** plots the variation of electrolyte pH at both applied current investigated. It was observed that the bulk pH decreased the most using carbon paper as cathode regardless of applied current. Under the polarization of 20 mA using carbon paper (**Fig. VIII.5(a**)), the bulk pH decreased to an average pH value of 5.12 before it dropped to 3.38 after 4 h and further down to 3.12 after 5 h of electrolysis. This explained why there was a re-dissolution of Mg and Ca precipitates starting from 3 h of electrolysis in the case of 20 mA. Regarding the pH evolution at higher applied current (**Fig. VIII.5(b**)), it dropped rapidly to 3.76 after just 1 h of electrolysis, before settling at 3.56 at the end of the process. Similarly, this could explain the re-dissolution of Ca as well as quicker depletion of TIC after 1 h of electrolysis when 200 mA was applied. From these trends of bulk pH, it could be summarized that the carbonate loss in case of carbon paper was principally due to the conversion of  $HCO_3^-$  to  $H_2CO_3$  (pK<sub>A</sub> = 6.3 [16]) and then to  $CO_2$  given the low pH environment (pH = 3.56) obtained using carbon paper. This was particularly true in the case of 20 mA where TIC was depleted by 94% but  $Ca^{2+}$  only precipitated by 11% (before re-dissolution). While applying 200 mA,  $Ca^{2+}$  quickly precipitated with  $CO_3^{2-}$  during the first hour of electrolysis before the pH environment got too low that it induced re-dissolution of  $CaCO_3$  after 1 h of electrolysis.

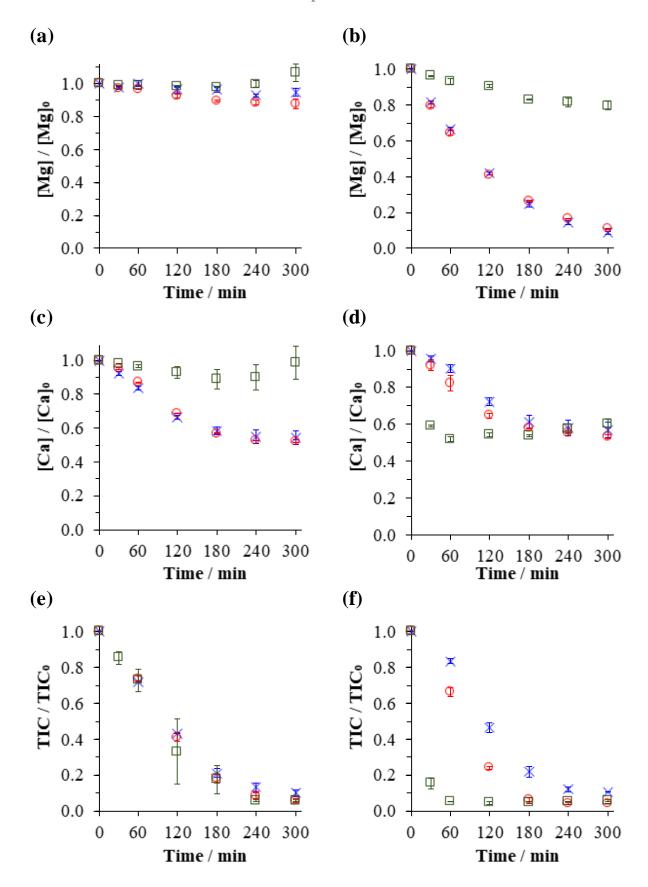
The evolution of bulk pH when stainless steel and graphite were used as cathode are also depicted in Fig. VIII.5. It can be observed that, regardless of the value of applied current, the electrolytic bulk pH decreased in the increasing order of stainless steel > graphite > carbon paper. Comparing the porosity of the three cathode materials, it increased in the increasing order of stainless steel < graphite < carbon paper. Another particularity using porous electrode as cathode is that, it has been reported that the porous nature of the electrode would induce a retardation of OH- diffusion flux from the porous cathode into the bulk [17]. As a consequence, more accumulation of OH in porous cathode due to the slower OH- diffusion led to higher local pH on porous cathode. In the same time, the precipitation reactions near the cathode surface consumed OH<sup>-</sup> [18, 19]. This consumption of OH<sup>-</sup> combined with its slower diffusion into the bulk could have led to higher uncombined H<sup>+</sup> concentration in the bulk, which resulted in lower pH value. This could be the reason why lower bulk pH was obtained using carbon paper as cathode, followed by graphite and stainless steel. Moreover, the higher porosity of carbon paper could have led to a higher accumulation of OH- in the pores, which could rapidly convert HCO<sub>3</sub><sup>-</sup> to CO<sub>3</sub><sup>2</sup>- for its reaction with Ca<sup>2+</sup> to form CaCO<sub>3</sub>. Consequently, rapid decrease of Ca<sup>2+</sup> concentration was noticed (Fig. VIII.4(d)). Following similar principle, the porous nature of graphite plate led to slightly higher local OH- concentration on graphite cathode relatively to the stainless steel plate which resulted in slightly higher CaCO<sub>3</sub> electro-precipitation on graphite as compared to stainless steel (Figs. VIII.4(d) and VIII.4(f)).

Contrastingly, the slow electro-precipitation of Mg(OH)<sub>2</sub> on carbon paper in comparison to graphite and stainless steel, came down to the applied cathode potential on carbon paper at 200 mA. It has been shown in literature that the precipitation of Mg(OH)<sub>2</sub> was highly dependent on applied current [20-23], or more precisely to the cathode potential [15]. It was noticed in Section VIII.3.1.1 that when 200 mA was applied, -1.54 V/Ag-AgCl was measured on stainless steel, -1.72 V/Ag-AgCl was measured on graphite, whilst -1.40 V/Ag-AgCl was measured on carbon paper (Fig. VIII.2). This higher applied cathode potential led to lower intensity of OH<sup>-</sup> production from the reduction reactions of dissolved O<sub>2</sub> and water (Eqs. (II.9) and (II.10)). As a result, slower Mg(OH)<sub>2</sub> electro-precipitation was apparent with carbon paper.

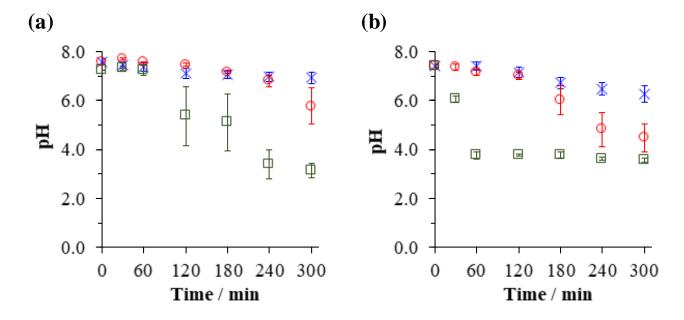
Due to differences that porosity could bring towards the mechanism of electro-precipitation, electro-oxidation experiments to treat RW effluent using porous (carbon paper) and non-porous cathode (stainless steel) are compared in the next sub-section.



**Fig. VIII.3**. Photographs of mineral electro-precipitation occurring on (a) stainless steel, (b) graphite and (c, d) carbon paper cathodes after 5 h of electrolysis.  $d_{\text{elec}}$ : 500  $\mu$ m.  $I_{\text{app}}$ : 200 mA.



**Fig. VIII.4**. Concentrations of Mg<sup>2+</sup>, Ca<sup>2+</sup> and TIC during the electrolysis at 20 mA ((a), (c), (e)) and 200 mA ((b), (d), (f)).  $d_{\text{elec}}$ : 500 µm, (×): stainless steel, (O): graphite and ( $\square$ ): carbon paper cathode. Electrolyte initially contained 150 mg L<sup>-1</sup> Ca<sup>2+</sup>, 5 mg L<sup>-1</sup> Mg<sup>2+</sup> and 60 mg-C L<sup>-1</sup> of TIC.



**Fig. VIII.5**. Evolution of bulk pH during the electrolysis at (a) 20 mA and (b) 200 mA. Cathode material: (★): stainless steel, (○): graphite and (□): carbon paper.

# VIII.3.1.3. Electro-precipitation on porous cathode during electro-oxidation of reclaimed wastewater

In this sub-section, the role of cathode porosity on the occurrence of electro-precipitate was further evaluated by treating RW effluent using an electro-oxidation process. The deposit formed on carbon paper (porous) and stainless steel (non-porous) along with the fate of inorganic and organic species in the treated media were quantified.

Figures VIII.6 and VIII.7 plot the evolution of inorganic and organic species during the electro-oxidation of RW when carbon paper and stainless steel served as cathode. From Fig. VIII.6(a), similar trend observed in BE was reproduced. Mg precipitated significantly on stainless steel cathode (93.0 ± 3.1%) but sluggishly on carbon paper (22.2%), which was ascribed to higher applied cathode potential on carbon paper as discussed previously (Section VIII.3.1.2). Moreover, similar conclusion could be drawn during the electrolysis of BE according to Figs. VIII.6(b) and VIII.6(c). There was a rapid CaCO<sub>3</sub> deposition on porous carbon paper (63.4% of Ca and 88.4% of carbonate precipitation after 2 h of treatment) relatively to that occurring on stainless steel. Again, similar reasoning in relation to the retardation of OH<sup>-</sup> diffusion to the bulk could be noticed. Thus, the accumulation of OH<sup>-</sup> in porous cathode remains valid. However, slightly more Ca precipitation was observed on stainless steel than on carbon paper at the end of electro-oxidation treatment, i.e. 70.0% and 64.4% on stainless steel and carbon paper, respectively. Only slight re-dissolution of CaCO<sub>3</sub> occurred in the carbon paper configuration in the final hour of treatment. This was supposed to be due to a better buffering capacity of the wastewater effluent. It can be demonstrated by the evolution of bulk pH

during the electrolysis as depicted in **Fig. VIII.6(d)**. When carbon paper was used as cathode, the bulk pH equaled to 6.42 after 3 h of treatment. Then the pH dropped down to 4.37, when the electrolysis ended. In comparison, the bulk pH was equivalent to neutral pH (7) when stainless steel was used. **Figure VIII.6(e)** shows the evolution of phosphates during the electro-oxidation processes. It was noticed that the formation of calcium phosphate took place independently of the porosity of cathode. It has been reported that the deposition of Ca<sub>3</sub>(PO<sub>4</sub>)<sub>2</sub> was facilitated on cathode with larger surface area under identical applied current since it offered higher reactive surface [24]. Nevertheless, that enhancement could not be identified under our operating conditions due to low phosphate content. Then, in agreement with the TOC evolution illustrated in **Fig. VIII.6(f)**, marginally better mineralization took place with stainless steel on cathode (28.3% vs. 16.5% using carbon paper).

**Figure VIII.7** illustrates the evolution of N and Cl species during the electro-oxidation process using carbon paper and stainless steel as cathode. **Figure VIII.7(a)** shows the concentration of NO<sub>3</sub><sup>-</sup> during the treatment. Using stainless steel cathode, NO<sub>3</sub><sup>-</sup> was only reduced by 9% after 1 h of electrolysis before its concentration increased progressively till the end of treatment. Meanwhile, when carbon paper replaced the stainless steel as cathode, NO<sub>3</sub><sup>-</sup> concentration continuously decreased over the course of electro-oxidation reaching 95.6% of removal after 5 h of electrolysis. In conjunction with the evolution of NH<sub>4</sub><sup>+</sup> concentration illustrated in **Fig. VIII.7(b)**, in which NH<sub>4</sub><sup>+</sup> increased over the course of treatment, it could be deduced that the NO<sub>3</sub><sup>-</sup> was reduced to yield NH<sub>4</sub><sup>+</sup> [25-28]. The reductions of NO<sub>3</sub><sup>-</sup> to NH<sub>4</sub><sup>+</sup> and N<sub>2</sub> (**Eqs. (VI.9)-(VI.11)** and **Eq. (VIII.3)**) were fast and they led to efficient elimination of NO<sub>3</sub><sup>-</sup> from the wastewater effluent.

$$NO_3^- + 10H^+ + 8e^- \rightarrow NH_4^+ + 3H_2O$$
 (VI.9)

$$NO_3^- + 6H^+ + 5e^- \rightarrow \frac{1}{2}N_2 + 3H_2O$$
 (VI.10)

$$NO_3^- + 2H^+ + 2e^- \rightarrow NO_2^- + H_2O$$
 (VI.11)

$$NO_3^- + 10H^+ + 8e^- \leftrightarrow NH_4^+ + \bullet OH + 3H_2O$$
 (VIII.3)

However, this evolution of NH<sub>4</sub><sup>+</sup> concentration was unexpected (**Fig. 7(b)**). Chloride was initially present in the RW ( $N_{Cl}$ -/ $N_{NH_4}$ +  $\geq$  100), surpassing the break point of 7.6 [29], but the NH<sub>4</sub><sup>+</sup> did not react with mediated chlorine oxidants to produce chloramines (**Eqs. (VII.26)-(VII.28)**), NO<sub>3</sub><sup>-</sup> (**Eq. (VII.30)**) and N<sub>2</sub> (**Eq. (VII.31)**).

$$HClO + NH_4^+ \rightarrow NH_2Cl + H_2O + H^+$$
 (VII.26)

$$NH_2Cl + HClO \rightarrow NHCl_2 + H_2O$$
 (VII.27)

$$NH_2Cl + HClO \rightarrow NCl_3 + H_2O$$
 (VII.28)

$$NH_4^+ + 4HClO \rightarrow NO_3^- + H_2O + 6H^+ + 4Cl^-$$
 (VII.30)

$$2NH_4^+ + 3HClO \rightarrow N_2 + 3H_2O + 5H^+ + 3Cl^-$$
 (VII.31)

Consequently, NH<sub>4</sub><sup>+</sup> concentration increased over the course of electro-oxidation, unlike the case using stainless steel. This finding could be explained by a low production of free chlorine (0.43 mg-Cl<sub>2</sub> L<sup>-1</sup>) when carbon paper was used compared with the stainless steel cathode (27.5 mg-Cl<sub>2</sub> L<sup>-1</sup>) (**Fig. VIII.7(c**)). Despite of that, the Cl<sup>-</sup> content of the effluent decreased by 18% more using carbon paper compared with stainless steel cathode (**Fig. VIII.7(d**)). In the meantime, the ClO<sub>3</sub><sup>-</sup> and ClO<sub>4</sub><sup>-</sup> concentrations increased. 179 mg-ClO<sub>3</sub><sup>-</sup> L<sup>-1</sup> and 11 mg-ClO<sub>4</sub><sup>-</sup> L<sup>-1</sup> were measured under carbon paper setup, whereas 193 mg-ClO<sub>3</sub><sup>-</sup> L<sup>-1</sup> and 4.7 mg-ClO<sub>4</sub><sup>-</sup> L<sup>-1</sup> were dosed under stainless steel configuration. Marginally lower ClO<sub>3</sub><sup>-</sup> by 7.3% and 2.3 times more ClO<sub>4</sub><sup>-</sup> were produced using carbon paper. It could be deduced that when porous carbon paper was used as cathode, the production of free chlorine via **Eqs. (VII.23)-(VII.25)** were much slower than the decomposition of free chlorine through **Eqs. (VII.37)-(VII.40)**, **Eqs. (VII.43)-(VII.44)**, **Eqs. (VII.53)-(VII.54)** and (**VII.56)-(VII.57)** as well as **Eqs. (VII.60)** and (**VII.66**).

$$Cl^- \rightarrow Cl_2 + 2e^-$$
 (VII.23)

$$Cl_2 + H_2O \to HClO + H^+ + Cl^-$$
 (VII.24)

$$\frac{1}{2}Cl_2 + H_2O \to HClO + H^+ + e^- \tag{VII.25}$$

$$6HClO + 3H_2O \rightarrow 2ClO_3^- + 4Cl^- + \frac{3}{2}O_2 + 12H^+ + 6e^-$$
 (VII.37)

$$HClO + 2ClO_2 + H_2O \rightarrow 2ClO_3^- + HCl + 2H^+$$
 (VII.38)

$$HClO \rightarrow ClO_3^- + HCl + Cl^-$$
 (VII.39)

$$HClO + H_2O \rightarrow ClO_2^- + 3H^+ + 2e^-$$
 (VII.40)

$$\frac{1}{2}Cl_2 + 3H_2O \to ClO_3^- + 6H^+ + 5e^-$$
 (VII.43)

$$3Cl_2 + 3H_2O \rightarrow ClO_3^- + 5Cl^- + 6H^+$$
 (VII.44)

$$Cl_2 + 4H_2O \rightarrow 2ClO_2 + 8e^-$$
 (VII.53)

$$\frac{1}{2}Cl_2 + 2H_2O \to HClO_2 + 3H^+ + 3e^-$$
 (VII.54)

$$HClO + H_2O \rightarrow HClO_2 + 2H^+ + 2e^-$$
 (VII.56)

$$HClO_2 \rightarrow ClO_2 + H^+ + e^-$$
 (VII.57)

$$HClO + \bullet OH \rightarrow \bullet ClO + H_2O$$
 (VII.60)

$$ClO^{-} + \bullet OH \rightarrow ClO_{2}^{-} + H^{+} + e^{-}$$
 (VII.66)

Moreover, the formation of ClO<sub>3</sub><sup>-</sup> would not only be due to the fast oxidation of HClO/ClO<sup>-</sup> and Cl<sub>2</sub> to ClO<sub>3</sub><sup>-</sup> (Eqs. (VII.37)-(VII.39)) and (Eqs. (VII.43)-(VII.44)) respectively, but also from direct oxidation of Cl<sup>-</sup> to ClO<sub>3</sub><sup>-</sup> (Eq. (VII.42)):

$$Cl^{-} + 3H_{2}O \rightarrow ClO_{3}^{-} + 6H^{+} + 6e^{-}$$
 (VII.42)

Judging from higher concentration of  $ClO_4^-$  under the carbon paper compared with the stainless steel setup, it could also be concluded that the transformation of  $ClO_4^-$  from  $ClO_3^-$  (**Eqs.** (**VII.45**) and (**VII.68**)) were faster in the porous condition. It therefore contributed to higher  $ClO_4^-$  concentration and in the same time lower  $ClO_3^-$  content in the treated media (**Fig. VIII.7(e**) vs. **Fig. VIII.7(f**)).

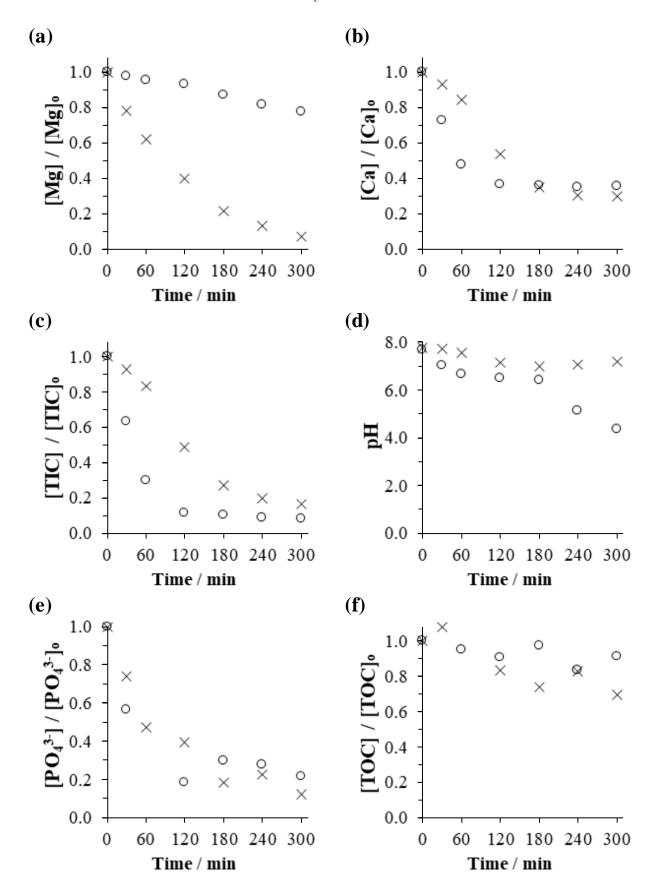
$$ClO_3^- + H_2O \rightarrow ClO_4^- + 2H^+ + 2e^-$$
 (VII.45)

$$ClO_3^- + \bullet OH \to ClO_4^- + H^+ + e^-$$
 (VII.68)

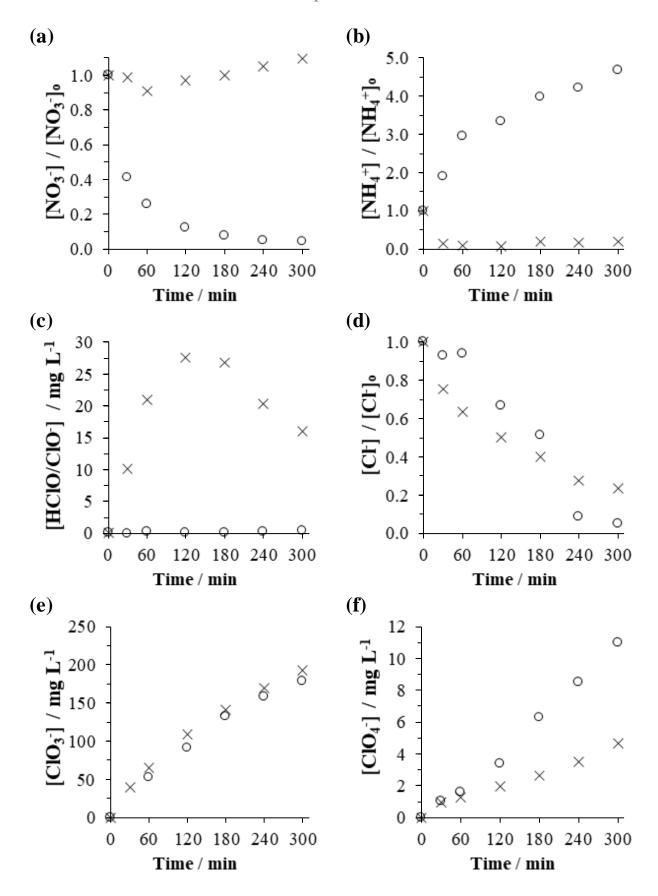
Figure VIII.8 depicts the SEM images of precipitate formed on the porous carbon paper after the 5 h-electrolysis of BE and RW at 200 mA. SEM could only be done on the carbon paper because stainless steel cathode could not be cut into the form of the SEM sampler for the analysis without affecting its surface. The morphologies of the raw carbon paper (as control) at different magnifications are provided in Figs. VIII.8(a) and VIII.8(b). With the benefit of EDX technique coupled to the SEM, 96.75 wt% of C was detected on the unused carbon paper sample with only 0.06 wt% Mg and 0.14 wt% Ca. After 5 h of electrolysis of BE (Figs. VIII.8(c) and VIII.8(d)), it was observed that CaCO<sub>3</sub> precipitated under cubic formed crystals typical of calcite allotropy. It was in agreement with the reports across literature given that the applied cathode potential (-1.40 V/Ag-AgCl) was in the region of reduction of water [30, 31]. It has also been reported that, in the presence of Mg<sup>2+</sup>, the growth of calcite crystals was slowed down and CaCO<sub>3</sub> would preferentially take aragonite allotropy involving needle-like crystals [20, 32-35]. However, the Mg<sup>2+</sup> concentration in our investigated system was too low to provoke significant change in calcite crystallography. Elementary analysis provided by EDX indicated that 1.23 wt% Mg and 40.39 wt% Ca was quantified on the carbon paper. The C wt% deposited on carbon paper that might belong to carbonate precipitate could not be precisely quantified due to the interference of C that might come from the carbon paper itself. In addition, Figs. VIII.8(c) and VIII.8(d) illustrate the morphology of electro-precipitate occurred after the electro-oxidation treatment of RW. It was noticed that in the presence of numerous ions and organics, the cubic feature of calcite allotropy slightly perished. Some part of the CaCO<sub>3</sub> deposition took irregular crystal form which might be a sign of co-precipitation with organic matter and/or phosphates as suggested by several authors [19, 33, 34, 36-38]. Further analysis using EDX indicated that 64.3 wt% of Ca and 0.90 wt% of Mg were present on the sampled carbon paper. About 0.61 wt% P was quantified compared to 0.08 wt% of P present on unused carbon paper control. However, the analyzed weight percentage of phosphorus presented high source of error since the concentration of PO<sub>4</sub><sup>3</sup>- was very low relatively to that of Ca<sup>2+</sup>. Moreover, the calcium phosphate deposition could take place anywhere on the 50 cm<sup>2</sup> geometric surface area of the carbon paper, i.e., 175 cm<sup>2</sup> electroactive surface area. This made difficult the detection of P within the material.

In summary, the use of highly porous cathode did not actually bring positive impact towards the minimization of electro-precipitation. This was due to (1) quicker CaCO<sub>3</sub> electro-precipitation that could occur when carbon paper served as cathode, (2) higher formation of ClO<sub>3</sub><sup>-</sup> and ClO<sub>4</sub><sup>-</sup> during the treatment, in addition to (3) the acidic environment produced at the end of the electrochemical process. Quick formation of CaCO<sub>3</sub> has to be avoided as it could rapidly passivate the cathode surface.

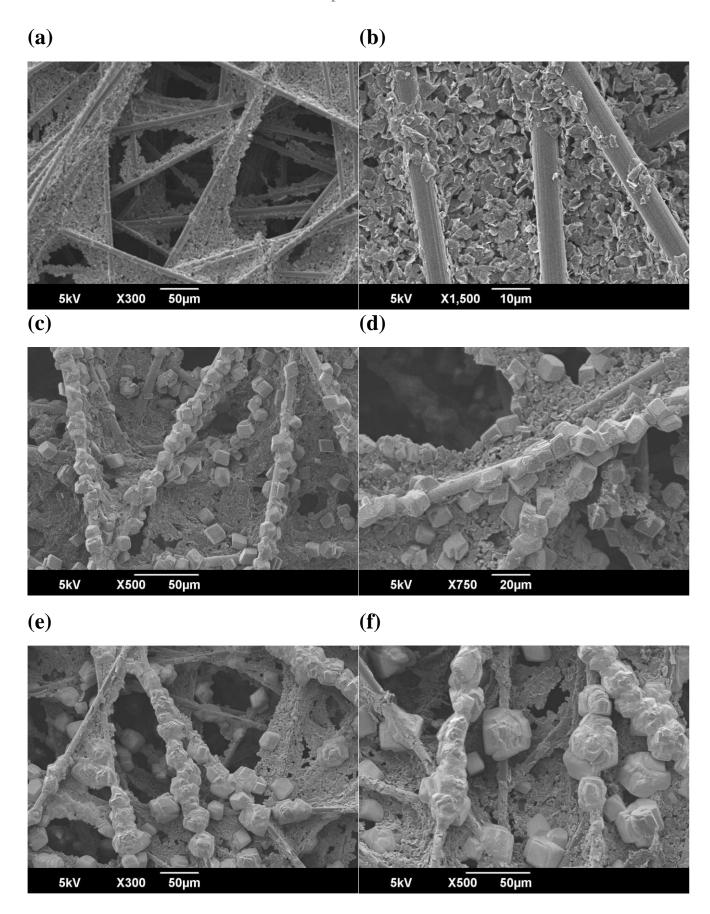
ClO<sub>3</sub><sup>-</sup> and ClO<sub>4</sub><sup>-</sup> are the well-known nocuous chlorinated byproducts and their formation is undesirable from the regulatory perspective. The acidic post-treatment condition might be favorable to re-dissolve the electro-precipitate formed on cathode surface, but it is not an efficient way to undo the passivated depositions. Most of the deposit remained on carbon paper despite minor re-dissolution into the bulk. Furthermore, the acidic environment would call for an additional treatment to neutralize the treated media before it could be released back into the environment.



**Fig. VIII.6**. Concentrations of (a) Mg<sup>2+</sup>, (b) Ca<sup>2+</sup>, (c) TIC, (d) effluent bulk pH, (e) PO<sub>4</sub><sup>3-</sup> and (f) TOC during the electro-oxidation treatment of RW at 200 mA applied current on (★) stainless steel and (O) carbon paper cathode. The properties of RW are given in **Table VII.1** of **Section VII.2**.



**Fig. VIII.7**. Concentrations of (a) NO<sub>3</sub>-, (b) NH<sub>4</sub>+, (c) HClO/ClO-, (d) Cl-, (e) ClO<sub>3</sub>- and (f) ClO<sub>4</sub>-during the electro-oxidation treatment of RW at 200 mA applied current on (×) stainless steel and (O) carbon paper cathode. The properties of RW are given in **Table VII.1** of **Section VII.2**.



**Fig. VIII.8**. SEM of unused carbon paper (control) ((a), (b)) and electro-precipitate formed on carbon paper after 5 h electrolysis of BE ((c), (d)) and RW ((e), (f)). *I*<sub>app</sub>: 200 mA.

# VIII.3.2. Role of H<sub>2</sub> evolution overpotential: synergistic effect of interelectrode distance and cathode material to reduce electro-precipitation

It has been demonstrated across **Section VIII.3.1** that by using a more porous cathode, it works in favor of mineral electro-precipitation. It corroborates the fact that porous cathode offers high electroactive surface, which enhances chemical and electrochemical reactions. As a consequence, if the formation of electro-precipitation is intended, porous cathode is the go-to option. Contrastingly, if the depreciation of mineral electro-precipitation is sought after, which is rather the case with the EAOPs in order to increase their life span, using cathode with high porosity might yield antagonistic consequences.

In this section, another property of carbonaceous material has been explored. Carbonaceous material such as carbon felt, graphite felt, vitreous carbon and graphite rod are known to possess higher overpotential for water reduction [7]. Consequently, wider potential window under cathodic range prior to H<sub>2</sub> evolution is achieved. That is the reason why carbonaceous materials have been receiving specific attention for a variety of applications (e.g. Fenton-based technology [28, 39-42], non-Fenton catalytic degradation of organics [43, 44], electrosorption [45] as well as fuel cells and batteries [3, 46, 47]) benefiting both their lower cost and environmentally-friendly properties. Moreover, it has been shown in Chapter 6 that, the electro-precipitation occurs with concomitant HER [20, 23, 32, 48, 49] which results in deposit detachment from cathode surface [23, 48, 50, 51]. In that regards, carbonaceous cathode is of no exemption. Due to determinant role of concurrent gas evolution taking an effect on electro-precipitation formation, the overpotential of H<sub>2</sub> evolution offered by a carbonaceous material was further explored. Graphite plate, a low porosity carbonaceous electrode (~15%) was used as cathode to meet the objective. The kinetics of electro-precipitation taking place on graphite together with stainless steel plate were compared under both submillimetric and millimetric configurations to inspect the inter-relation between interelectrode gap and H<sub>2</sub> overpotential. The mechanistic understanding at the cathode/electrolyte interface is further deepened through electrochemical impedance analysis presented in Section VIII.3.2.2.

# VIII.3.2.1. Electro-precipitation on stainless steel and graphite at different interelectrode distances

**Figure VIII.9** shows the evolution of  $Mg^{2+}$ ,  $Ca^{2+}$  and TIC deposited during the electrolysis at 200 mA, when either graphite or stainless steel plates were used as cathode. From the plot of  $Mg^{2+}$  concentration evolution in **Fig. VIII.9(a)**, it was observed that more  $Mg^{2+}$  electro-precipitated using 500 μm  $d_{elec}$  in comparison with 3000 μm setup. Using 500 μm, comparable  $Mg^{2+}$  electro-

precipitation was noticed between stainless steel (91.0  $\pm$  1.8%) and graphite (89.1  $\pm$  2.2%). When 3000 µm  $d_{\rm elec}$  was equipped, Mg<sup>2+</sup> precipitated by 81.1  $\pm$  0.2% and 82.6  $\pm$  2.8% on stainless steel and graphite, respectively. Under the 100  $\mu$ m setup, a different behavior was noticed. 85.6  $\pm$  2.6% of Mg<sup>2+</sup> precipitated on stainless steel which was comparable to that evaluated using 500 µm setup considering the standard deviations. Contrastingly, only 6.3  $\pm$  0.1% of Mg<sup>2+</sup> was deposited on graphite cathode. In addition, the evolution of Ca<sup>2+</sup> concentration is depicted in Fig. VIII.9(b). Reproducible trends of Ca<sup>2+</sup> deposition, relatively to the aforementioned Mg<sup>2+</sup> deposition, were noted for the reactor configurations with both 500 μm and 3000 μm interelectrode gaps. More Ca<sup>2+</sup> precipitation occurred under the configuration of 500 µm compared to that of 3000 µm. Under the 500 µm configuration,  $Ca^{2+}$  deposited by 43.2  $\pm$  5.6% on stainless steel, whereas it precipitated slightly more on graphite cathode (i.e., by 46.7  $\pm$  1.2%). In comparison with the 3000  $\mu$ m  $d_{\rm elec}$ , Ca<sup>2+</sup> precipitated by  $36.2 \pm 1.6\%$  on stainless steel versus  $40.0 \pm 3.3\%$  on graphite. Surprisingly, when  $100 \, \mu m \, d_{\text{elec}}$  was used, a conflicting behavior was noticed again. It corresponded to the configuration in which CaCO<sub>3</sub> electro-precipitate occurred the most at stainless steel cathode (55.1  $\pm$  2.2%), compared with graphite electrode (9.0  $\pm$  1.2%). The evolution of TIC concentrations illustrated in Fig. VIII.9(c) followed similar deduction drawn for the fate of Ca<sup>2+</sup>. TIC decreased the most under the configuration of 100  $\mu$ m  $d_{\rm elec}$  with stainless steel as cathode. Comparatively, more carbonate deposition occurred under the 500  $\mu$ m  $d_{\rm elec}$  setup as compared to the 3000  $\mu$ m, with marginally more electro-precipitation evaluated on graphite cathode in comparison to stainless steel for both  $d_{\text{elec}}$ .

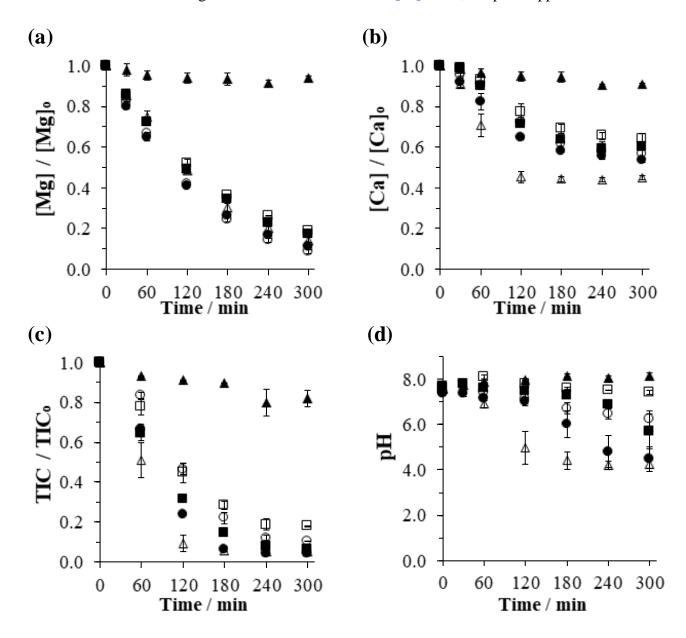
To recapitulate, more electro-precipitates was formed under the 500  $\mu$ m configuration (microfluidic) in comparison to that of 3000  $\mu$ m (millimetric) and slightly more electro-precipitation took place on graphite cathode relatively to the stainless steel. The worst electro-precipitation occurred using 100  $\mu$ m (microfluidic)  $d_{\rm elec}$  but only when stainless steel was used as cathode. Contrastingly, the slowest electro-precipitation was found also using the 100  $\mu$ m  $d_{\rm elec}$  but with graphite on cathode. Upon these results, two general conclusions could be assumed: (1) interelectrode distance played determining role towards the formation of electro-precipitation and (2) under very particular operating condition (i.e., submillimetric  $d_{\rm elec}$  equals or lower than 100  $\mu$ m), graphite properties could help to reduce the kinetics of deposition unlike any other experimental conditions.

To further understand this disparity of electro-precipitation in function of cathode material as well as  $d_{\rm elec}$ , the polarization curves in cathodic region on stainless steel and graphite at 100, 500 and 3000 µm gaps were performed (**Fig. VIII.10**). According to **Fig. VIII.10(a)**, it was observed that the HER overpotential was lower using graphite cathode in comparison to stainless steel at 500 µm  $d_{\rm elec}$ . As a result, when 200 mA current was applied (i.e., 4 mA cm<sup>-2</sup>), approximately -1.54 V/Ag-AgCl was measured on stainless steel, whilst -1.72 V/Ag-AgCl was quoted on graphite. Similarly, according to

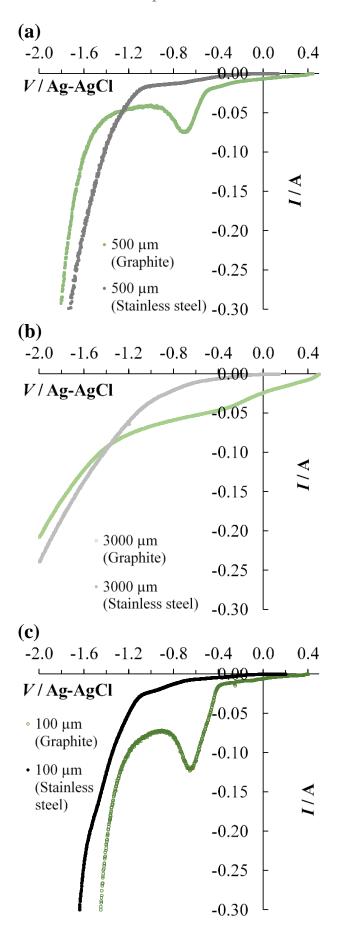
Fig. VIII.10(b), when the cell was used with 3000  $\mu$ m  $d_{\text{elec}}$ , the H<sub>2</sub> overpotential was also lower at 200 mA on graphite (-1.97 V/Ag-AgCl) than on stainless steel (-1.91 V/Ag-AgCl) cathodes. On one hand, the significantly lower applied cathode potential (into the water reduction region) measured under the macrometric reactor setup (3000 µm) could explain the lesser degree of electroprecipitation under this configuration. Intense competitive H<sub>2</sub> evolution on cathode surface reduced the reactive site for precipitating reactions and in the same time induced partial deposit detachment from the cathode surface. This observation follows the deduction made in **Chapter 6**, where the role of  $d_{\text{elec}}$  was thoroughly discussed and modeled. On the other hand, it was also noted that the applied cathode potential was marginally lower on graphite than on stainless steel (Figs. VIII.10(a) and VIII.10(b)). This led to slightly enhanced OH<sup>-</sup> ions production from the water reduction reaction on graphite (Eq. (II.10)). As a consequence, there was a marginally higher driving force of thermodynamic shift of Eqs. (II.11)-(II.13) to the right, in the direction of mineral scaling production. Furthermore, it has been demonstrated that graphite inherits porous characteristic. It offered slightly enhanced electroactive surface area for electronic transfers (85 vs. 80 cm<sup>2</sup> of stainless steel (Section VIII.3.1.1)), which could lead to better electrochemical and/or chemical reactions. Local pH on graphite was also slightly higher than that achieved on stainless steel due to slight retardation of OHdiffusion into the bulk. This led to better CO<sub>3</sub><sup>2-</sup> conversion for CaCO<sub>3</sub> precipitation reaction in the case of graphite. These arguments could justify the fact that more deposition was quantified on graphite rather than on stainless steel at 3000  $\mu$ m and 500  $\mu$ m  $d_{\rm elec}$ . On the contrary, reversal trend was noticed in the case where the microfluidic cell was equipped with 100  $\mu$ m  $d_{\rm elec}$ . In accordance with Fig. VIII.10(c), it was observed that the HER overpotential was higher on graphite rather than on stainless steel cathode. Consequently, the cathode potential was measured at -1.33 V/Ag-AgCl on graphite against -1.45 V/Ag-AgCl on stainless steel when 200 mA was applied. Subsequently, less OH ions were produced on graphite, which significantly lowered the kinetics of Mg(OH)<sub>2</sub> and CaCO<sub>3</sub> deposition. Therefore, this finding might explain the contrasting observations made between the two cathode materials when 100 µm delec was adopted. This was the only condition where the carbonaceous graphite would substantially minimize the formation of mineral scaling.

The difference in the occurrence of mineral scaling on graphite and stainless steel inside microfluidic and millimetric reactors could be further emphasized by the evolution of bulk pH during the electrolysis (**Fig. VIII.9(d)**). The decreasing pH value correlated with the OH<sup>-</sup> ion consumption by **Eqs. (II.11)** and (**II.12**) to form mineral precipitates. The more pH decreased, the more electroprecipitate formed on cathode surface, which was in agreement with the curves depicted in **Figs. VIII.9(b)** and **VIII.9(c)**. Only bulk pH corresponding to the microfluidic setup (100 µm) using graphite cathode increased over time. It corroborated with the fact that this was the only configuration where sluggish mineral scaling was observed (**Fig. VIII.9**). The pH evolution depicted in **Fig.** 

**VIII.9(d)** again confirmed the accumulation of OH<sup>-</sup> near porous cathode when graphite was used, which resulted in increasing H<sup>+</sup> accumulation in the bulk [17]. Thus, the pH dropped.



**Fig. VIII.9**. Concentrations of (a) Mg<sup>2+</sup>, (b) Ca<sup>2+</sup>, (c) TIC and (d) bulk pH during the electrolysis of solution initially containing 150 mg L<sup>-1</sup> Ca<sup>2+</sup>, 5 mg L<sup>-1</sup> Mg<sup>2+</sup> and 60 mg-C L<sup>-1</sup> of TIC at 200 mA.  $d_{\text{elec}}$  of ( $\triangle$ ,  $\blacktriangle$ ): 100, ( $\bigcirc$ ,  $\bigcirc$ ): 500 and ( $\square$ ,  $\blacksquare$ ): 3000 μm were evaluated using ( $\blacktriangle$ ,  $\bigcirc$ ,  $\blacksquare$ ): graphite and ( $\triangle$ ,  $\bigcirc$ ,  $\square$ ): stainless steel cathode. Anode: BDD.



**Fig. VIII.10**. LSV of stainless steel and graphite plate cathodes when they were assembled in an electrochemical cell with (a) 500, (b) 3000 and (c) 100  $\mu$ m  $d_{\rm elec}$ . The potential was swept between 0.5 to -1.8 V/Ag-AgCl with a scan rate of 10 mV s<sup>-1</sup>. CE: BDD, RE: Ag-AgCl.

# VIII.3.2.2. Cathode/electrolyte interface study of electro-precipitation on graphite and stainless steel by electrochemical impedance technique

EIS analysis was performed on graphite and compared to stainless steel in order to characterize mineral electro-precipitation during the electrolysis at 100, 500 and 3000 µm interelectrode gaps. The evolutions of  $R_{\rm CT}$  and  $C_{\rm DL}$  using graphite cathode are plotted in Fig. VIII.11. It can be noticed that the values of  $R_{CT}$  appeared constant during the electrolysis for all three  $d_{elec}$  (Figs. VIII.11(a), VIII.11(c) and VIII.11(e)).  $R_{\rm CT}$  measured on graphite took the average values of 0.42  $\pm$  0.10, 0.88  $\pm$  0.05 and 2.96  $\pm$  0.10  $\Omega$  respectively, when 100, 500 and 3000  $\mu$ m  $d_{\rm elec}$  were used. Despite the formation of electro-precipitation in the configurations of 500 and 3000 µm, the graphite surface was also not fully passivated. This was in-trend with the conclusion obtained in Chapter 6, using stainless steel as cathode. As for the evolution of  $C_{DL}$  illustrated in Figs. VIII.11(b), VIII.11(d) and VIII.11(f), it was noticed that they rather increased during the electrolysis unlike the evolution of  $C_{\rm DL}$  on stainless steel cathode. It increased by 58, 76 and 37% in the respective order of 100, 500 and 3000 µm reactor configuration. This increase of  $C_{DL}$  values had no significant relation with the formation of electroprecipitate since it also increased when 100  $\mu$ m  $d_{elec}$  was used (a condition where the least mineral scaling took place). Instead, it has something to do with the porosity of graphite and will be discussed shortly when it is compared to the stainless steel cathode. With respect to the evolution of film resistance  $(R_{\rm F})$  formed on graphite during the electrolysis plotted in Fig. VIII.12, it was observed that  $R_{\rm F}$  increased during the first 2 h of electrolysis to 0.29  $\pm$  0.01 and 0.43  $\pm$  0.15  $\Omega$  for 500 and 3000  $\mu$ m setup, respectively. It was expected to measure higher  $R_F$  value using 500  $\mu$ m  $d_{elec}$  since marginally more scaling occurred at this distance, but the opposite trend was observed. Nonetheless, after 2 h electrolysis, reproducible trend was noticed on both interelectrode gaps in line with the kinetics of precipitation previously discussed in Section VIII.3.1.2 and Section VIII.3.2.1. R<sub>F</sub> values in both conditions decreased and it could be ascribed to the evolution of bulk pH in reference to Fig. VIII.9(d). The fact that the electro-precipitate had the possibility to re-dissolve into the solution made decrease the apparent resistance related to the precipitated film. This might be the reason why  $R_{\rm F}$ values decreased in both  $d_{\text{elec}}$  after 2 h of electrolysis as depicted in Fig. VIII.12.

Upon comparing the interfacial activities between graphite and stainless steel, some differences could be highlighted. The ratio of  $R_{\rm CT}$  and  $C_{\rm DL}$  values between graphite and stainless steel for all investigated  $d_{\rm elec}$  configurations are illustrated in **Fig. VIII.11**. It can be observed that  $R_{\rm CT}$  values on graphite were lower than those on stainless steel, i.e., averagely by 0.6-fold, 0.8-fold and 0.7-fold using 100, 500 and 3000  $\mu$ m, respectively in all conditions. This lower  $R_{\rm CT}$  value characterized by EIS on graphite relatively to that on stainless steel confirmed the similar deduction on  $R_{\rm CT}$  comparison in **Section VIII.3.1.1** using  $I_0$  method. On the contrary,  $C_{\rm DL}$  values were higher on graphite in

comparison to the values measured on stainless steel by an average factor of 2.7, 1.1 and 1.5 using 100, 500 and 3000  $\mu$ m, respectively. These differences were accounted for the porous properties of graphite. Due to the higher specific surface area of electrode/electrolyte interface using graphite (85 vs. 80 cm<sup>2</sup> for stainless steel, (Section VIII.3.1.1)), more active sites were available for electronic exchange at the interface. Consequently,  $R_{CT}$  decreased and  $C_{DL}$  increased. This finding was in agreement with the results documented by Da Silva et al. [52]. They found that  $C_{DL}$  value measured on porous metal oxide electrode was significantly reduced when the porosity of the electrode was decreased.

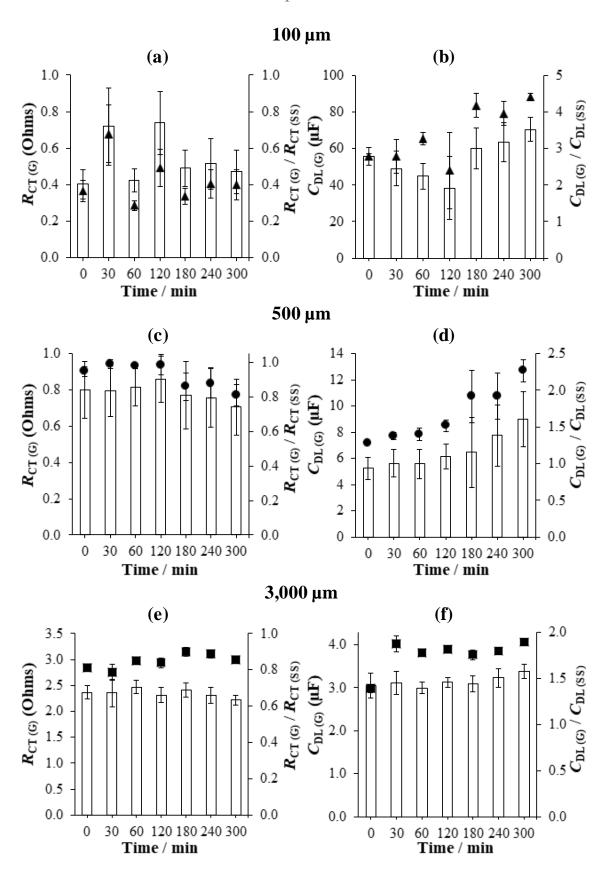
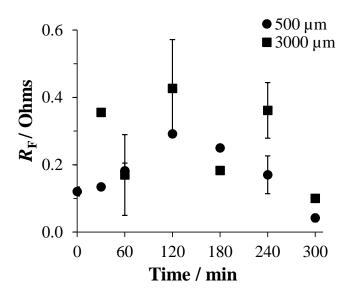


Fig. VIII.11. Evolutions of  $R_{\text{CT(G)}}(\triangle, \bullet, \blacksquare)$  and  $R_{\text{CT(G)}}/R_{\text{CT(SS)}}(\square)$  ((a), (c), (e)) as well as  $C_{\text{DL(G)}}(\triangle, \bullet, \blacksquare)$  and  $C_{\text{DL(G)}}/C_{\text{DL(SS)}}(\square)$  ((b), (d), (f)) during the electrolysis of solution containing Ca<sup>2+</sup> 150 mg L<sup>-1</sup>, Mg<sup>2+</sup> 5 mg L<sup>-1</sup> and TIC 60 mg L<sup>-1</sup>. ( $\triangle$ ): 100 µm ((a), (b)), ( $\bullet$ ): 500 µm ((c), (d)) and ( $\blacksquare$ ): 3000 µm ((e), (f)). G: graphite and SS: stainless steel.



**Fig. VIII.12**. Evolutions of  $R_F$  during the electrolysis of solution initially containing Ca<sup>2+</sup> 150 mg L<sup>-1</sup>, Mg<sup>2+</sup> 5 mg L<sup>-1</sup> and TIC 60 mg L<sup>-1</sup>. ( $\blacksquare$ ): 500 and ( $\blacksquare$ ): 3000  $\mu$ m. WE: graphite, CE: BDD and RE: Ag-AgCl.

#### VIII.4. Conclusions

The role of cathode materials was meticulously investigated in this chapter. Non-porous and non-carbonaceous stainless steel as well as carbonaceous graphite plate and carbon paper which possessed different porosities were compared. The occurring electro-precipitation on different cathode materials was studied under submillimetric configuration, a condition in which no electro-precipitation results have yet been reported. Moreover, the influence of cathode material, particularly the porosity was examined using real RW. Simultaneously, the fate of chlorinated and nitrogenous inorganic byproducts were followed and quantified. Electroactive surface area of stainless steel, graphite and carbon paper were determined via electrochemical method. A value of 80, 85 and 175 cm<sup>2</sup> was obtained for stainless steel, graphite and carbon paper, respectively. It indicated that different cathode materials offered different number of sites for electronic exchanges on top of having different porosities. Principle results obtained using different cathode materials are as follows:

- Mg(OH)<sub>2</sub> precipitated little on stainless steel, graphite and carbon paper at low applied current (20 mA) whilst CaCO<sub>3</sub> electro-precipitation was formed only on stainless steel and graphite cathode but not on carbon paper.
- Using higher applied current (200 mA), significant precipitation of Mg(OH)<sub>2</sub> was observed on stainless steel and graphite, but low precipitation on carbon paper. However, quick CaCO<sub>3</sub> deposition was noted on carbon paper. Meanwhile, slightly favored CaCO<sub>3</sub> electro-precipitation on graphite in comparison to stainless steel was observed and it was ascribed to higher local pH due to porous characteristic of graphite.
- When porous carbon paper and non-porous stainless steel were used to treat RW, comparable
  mechanism of electro-precipitation was noticed compared to the BE. Marginal less organics
  mineralization was achieved using carbon paper.
- Lower HER overpotential would only lead to more formation of electro-precipitation. It was due to enhanced OH<sup>-</sup> production that would consequently favor the mineral scaling. Slightly less mineral scaling could be achieved if the HER overpotential was very low into the water reduction region (where the competing H<sub>2</sub> evolution rigorously evolved on cathode surface).
- The role of HER overpotential was overshadowed by the additional rate limiting parameter, i.e., the  $E_{\rm C}$ . When stainless steel was used as cathode under the 100  $\mu$ m  $d_{\rm elec}$  setup, the most electroprecipitation took place. Contrasting, by replacing stainless steel with graphite, the slowest mineral scaling was noticed. It was due to the alteration in cathode potential value where the OH-production was less favorable.
- Using electrochemical impedance analysis, porous attribute of graphite was evidenced. Higher  $C_{\rm DL}$  was determined on graphite compared to stainless steel. Lower  $R_{\rm CT}$  measured on graphite

proved that better electronic transfers take place on graphite in comparison with stainless steel cathode.

In overall, due to rapid CaCO<sub>3</sub> deposition, high conversion rate of Cl<sup>-</sup> into ClO<sub>3</sub><sup>-</sup> and ClO<sub>4</sub><sup>-</sup> as well as acidic post-treatment condition, porous carbon paper was not a suitable cathode to be opted for if the minimization of electro-precipitation is sought. Even though carbonaceous material such as graphite presents an economical edge, its use only resulted in worse appearance of mineral scaling. Nonetheless, it has been proven herein that slow kinetic growth of deposit could however be devised using graphite or other materials under the condition where the applied cathode potential does not favor the OH<sup>-</sup> production.

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#### IX. General conclusions

#### IX.1. General overview

Submillimetric electrochemical reactors has bright potential to be used in the field of water/wastewater treatment. The micrometric  $d_{\rm elec}$  not only intensifies mass transfer at the core of the electrochemical reactor, but also enhances redox reactions in between electrodes. Significant drop in interelectrode gap reduces internal resistance lowering notably the power consumption of the cell. Motivated by genuine interest in the use of microfluidic in the environmental domain, a thorough literature survey (Chapter 2) proved that little attention has been given on the occurrence of mineral deposition in submillimetric features. Consequently, it has been the principal goal of the PhD thesis to characterize the kinetics of electro-precipitation in such a reactor design and to explore the influencing parameters around it. The parametric approach has been proven correct as insightful experimental data were exploited. Moreover, theoretical models on mineral scaling kinetics as well as on interfacial properties of scaled cathode/effluent have been proposed.

#### IX.2. Kinetic models of mineral scaling inside microfluidic reactors

Reliable mathematical models have been proposed in **Chapter 4** and **Chapter 6** to predict the formation of electro-precipitation during the electro-oxidation process. The models took into consideration the  $j_{app}$ , electrolyte composition (Mg<sup>2+</sup>, Ca<sup>2+</sup>, OH<sup>-</sup> and HCO<sub>3</sub><sup>-</sup>/CO<sub>3</sub><sup>2-</sup>) in the bulk as well as the  $d_{elec}$  by intermediary of concomitant evolving gas. Using the theoretical models, the experimental and model curves of precipitating Mg<sup>2+</sup> and Ca<sup>2+</sup> matched well. Moreover, the interfacial  $R_{CT}$  and  $C_{DL}$  properties at varying  $d_{elec}$  which were acquired experimentally using electrochemical impedance were also fitted under mathematic models. Consequently, the developed models would allow an estimation of the kinetics of mineral scaling as well as the values of  $R_{CT}$  and  $C_{DL}$  as function of large range of  $d_{elec}$  from submillimetric to millimetric ranges.

#### IX.3. Mass transfer evolution in submillmetric vs. millimetric configuration

Prior to the investigation of electro-precipitation occurring inside electrochemical reactors with varying  $d_{\rm elec}$ , the mass transfer evolution while varying the  $d_{\rm elec}$  was firstly looked into in **Chapter 5**. It has been demonstrated that  $k_{\rm m}$  significantly increased with decreasing  $d_{\rm elec}$ , particularly for interelectrode gap below 1000  $\mu$ m. It thus defined a quantitative boundary between microfluidic and macrofluidic behaviors. Furthermore, a theoretical model that correlates  $k_{\rm m}$  with a wide range of  $d_{\rm elec}$  (100 to 3000  $\mu$ m) was established. It would help an easy estimation of  $k_{\rm m}$  for a given electrochemical cell equipped with a large range of  $d_{\rm elec}$  working in parallel-plate mode. On top of that, mass transfer

correlation based on the Sherwood number in laminar regime was proposed. It took into consideration, not only fluid properties but also electrochemical reactor hydrodynamics and geometry for a convenient scale up or scale down application. Owing to the establishment of a  $k_{\rm m}$  model with  $d_{\rm elec}$ , its value could have been estimated when a  $d_{\rm elec}$  of 50  $\mu$ m was used. The determination of  $I_{\rm lim}$  for  $k_{\rm m}$  evaluation under this configuration was technically impossible due to the capacity limitation of the potentiostat.

### IX.4. Fundamental role of interelectrode distance and energetic performance

Once the mass transfer characterization of the electrochemical cell with varying  $d_{\rm elec}$  has been grasped, the research moved on to the role of  $d_{\text{elec}}$  on the formation of mineral electro-precipitation in Chapter 6. Table IX.1 and Fig. IX.1 illustrate the results of electro-precipitates formation and energetic performance during the electro-oxidation of a synthetic effluent under different  $d_{\text{elec}}$ configurations. They reflect the relationship of process efficiency and operational cost under different reactor configurations. From both Table IX.1 and Fig. IX.1, no remarkable trend was seen on the evolution of mass of  $Mg(OH)_2$  deposition as a function of  $d_{elec}$ . Meanwhile, insignificant quantity of CaCO<sub>3</sub> precipitation occurred using 50 µm due to non-faradaic conditions. It has been demonstrated in Chapter 6 that increasing  $d_{\text{elec}}$  resulted in decreasing  $E_{\text{C}}$  into water reduction region. The least parasitic HER took place under the configuration of 100 µm. Consequently, the most CaCO<sub>3</sub> electroprecipitated using the 100  $\mu$ m  $d_{\rm elec}$  (35.3 mg). As the  $d_{\rm elec}$  increased, the mass of CaCO<sub>3</sub> deposition decreased (down to 27.2 mg using 3000 µm setup) (refer **Table IX.1** and the relative trend plotted in Fig. IX.1. With regard to the power consumption (PC) during the electrolysis of 5 h across submillimetric into millimetric configurations, increasing trend of PC was evaluated in the increasing order of  $d_{\rm elec}$ . The lowest PC was evaluated at  $d_{\rm elec} = 50 \, \mu \rm m \, (0.02 \, kWh \, m^{-3})$  and it raised to 9.97 kWh m<sup>-3</sup> using 3000 μm (**Table IX.1** and **Fig. IX.1**). This increasing trend of power consumption was in close relation with the variation of cell voltage ( $\Delta U$ ) (Fig. IX.1).  $\Delta U$  was noticed to increase from 0.02 to 4.99 V when the reactor was used with  $d_{\rm elec}$  of 50 and 3000  $\mu m$  respectively. From the total amount of electro-precipitate (i.e., the sum of mass of Mg(OH)<sub>2</sub> and CaCO<sub>3</sub> deposit), specific power consumption for the formation of electro-precipitate (SPCP) during 5 h of electrolysis was evaluated. SPCP of 0.0117 and 0.1707 kWh g-precipitate<sup>-1</sup> were obtained under  $d_{\text{elec}} = 50$  and 3000  $\mu$ m setup respectively (Table IX.1). The increasing trend of SPCP (Fig. IX.1) was due to (1) increasing PC as well as (2) decreasing quantity of electro-precipitate in the ascending order of  $d_{\text{elec}}$ .

**Table IX.1**. Electro-precipitation and energetic performance during 5 h of electrolysis in basic effluent (BE) across investigated  $d_{\text{elec}}$ .

		· /	<u> </u>			
	50 µm	100 µm	250 μm	500 μm	1,000 µm	3,000 µm
Mg(OH) <sub>2</sub> (mg)	0.00	2.00	2.22	1.97	2.16	2.02
CaCO <sub>3</sub> (mg)	1.610	35.28	34.18	30.29	29.20	27.18
$\Delta U(\mathbf{V})$	0.0200	3.85	3.94	4.01	4.21	4.99
PC (kWh m <sup>-3</sup> )	0.0380	7.70	7.88	8.03	8.41	9.97
SPCP (kWh g-precip1)	0.01170	0.1032	0.1083	0.1287	0.1300	0.1707

**PC**: specific power consumption per volume of effluent treated and **SPCP**: specific power consumption per gram of electro-precipitate  $(Mg(OH)_2 + CaCO_3)$ .

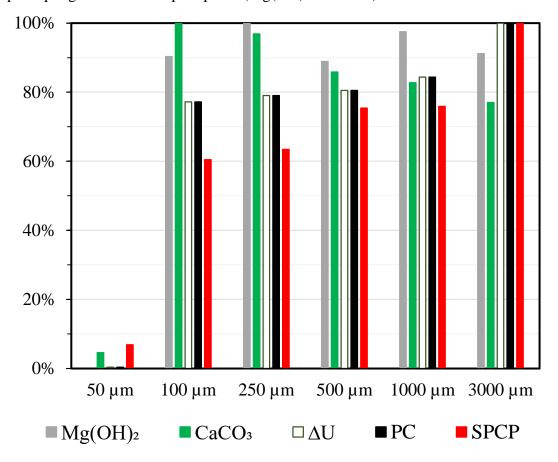


Fig. IX.1. Variation of Mg(OH)<sub>2</sub> and CaCO<sub>3</sub> electro-precipitation, cell voltage ( $\Delta U$ ), power consumption (PC) and specific power consumption per gram of precipitate (SPCP) as a function of  $d_{\rm elec}$ . The reported value of a given parameter was relative to the highest one obtained with a given  $d_{\rm elec}$  (Table IX.1).

# IX.5. Technical aspects of electro-oxidation treatment of reclaimed wastewater at different applied current densities

The study was pursued with the characterization of mineral deposition in the matrix having increasing degree of complexity, close to the composition of a real effluent of municipal WWTP, and ended up with comparison with RW itself in **Chapter 7**. In regard to the theme of the thesis, simultaneous (1)

electro-precipitation, (2) organics mineralization, (3) removal of target pharmaceutical micropollutant and (4) fate of inorganic byproducts were quantified in submillimetric setup, a configuration distinguishable from any other work ever documented on the topic. To reach the objective,  $d_{\rm elec}$  of 500 µm was selected to be in the microfluidic condition and in the same time having the least mineral scaling as deduced from Chapter 6. Table IX.2 summarizes some evaluation aspects of the electro-oxidation of RW spiked with 10 mg L<sup>-1</sup> tylosin at 0.8, 4 and 16 mA cm<sup>-2</sup>. The highest degree of mineralization (51.1%) after 5 h of treatment was achieved using the highest  $j_{app}$ (16 mA cm<sup>-2</sup>), whereas the lowest value (7.7%) corresponded to the lowest  $j_{app}$ , i.e., 0.8 mA cm<sup>-2</sup>. The mass of electro-precipitation tabulated in Table IX.2 corresponds to the total mass of Mg(OH)<sub>2</sub>, CaCO<sub>3</sub> and Ca<sub>3</sub>(PO<sub>4</sub>)<sub>2</sub> precipitates. Marginally lower degree of electro-precipitation was observed after 5 h at 0.8 (53.1 mg) than at 4 mA cm<sup>-2</sup> (56.6 mg), due to more OH<sup>-</sup> production on cathode in the latter. Further increase in  $j_{app}$  (16 mA cm<sup>-2</sup>) resulted in lower formation of electro-precipitation (47.1 mg) as the consequence of increasing competitive HER evolution on cathode surface. Next, harmful byproducts were quantified as the sum of ClO<sub>3</sub><sup>-</sup> and ClO<sub>4</sub><sup>-</sup> concentrations produced, expressed in mg-Cl<sub>2</sub> L<sup>-1</sup>. It was noticed that the highest value (151 mg-Cl<sub>2</sub> L<sup>-1</sup>) was obtained using 16 mA cm<sup>-2</sup>, whilst the lowest (2.9 mg-Cl<sub>2</sub> L<sup>-1</sup>) was yielded using the lowest  $j_{app}$ . In terms of power consumption performance during the 5-h electro-oxidation process, the highest PC was obtained by applying 16 mA cm<sup>-2</sup> (41.2 kWh m<sup>-3</sup>). On the other hand, the lowest PC was evaluated at 1.29 kWh m<sup>-3</sup> upon applying 0.8 mA cm<sup>-2</sup>. This trend was expected since the power consumption was proportionate to the applied current.

Total degradation of tylosin (100%) was obtained under the polarization of 4 and 16 mA cm<sup>-2</sup> after 20 and 1 min of treatment, respectively, whereas 95% tylosin removal was noted after 5 h of electrolysis at 0.8 mA cm<sup>-2</sup>. As a result, the specific power consumption for complete tylosin degradation (*SPCT*) was evaluated as function of not only the amount of tylosin removed (per gram of tylosin) but also of the applied current and the duration of the process. According to **Table IX.2**, the lowest *SPCT* (0.017 kWh g-Tyl<sup>-1</sup>) was evaluated at 16 mA cm<sup>-2</sup>, due to swift and total degradation of tylosin. Due to long treatment duration using 0.8 mA cm<sup>-2</sup>, the highest *SPCT* was calculated using this  $j_{app}$  (0.18 kWh g-Tyl<sup>-1</sup>), higher than *SPCT* at 4 mA cm<sup>-2</sup> by 2.4-fold (0.069 kWh g-Tyl<sup>-1</sup>). Similar approach was followed for the evaluation of specific power consumption for organic mineralization (*SPCM*) and electro-precipitation (*SPCP*). The extent of mineralization and electro-precipitation obtained after 5 h of electro-oxidation were used for the calculation of *SPCM* and *SPCP* respectively at the three investigated  $j_{app}$ . As expected, 0.8 mA cm<sup>-2</sup> gave out the lowest *SPCM* (0.495 kWh g-TOC<sup>-1</sup>) and *SPCP* (0.012 kWh g-precipitate<sup>-1</sup>), whereas the highest *SPCM* (3.01 kWh g-TOC<sup>-1</sup>) and *SPCP* (0.44 kWh g-precipitate<sup>-1</sup>) were evaluated using 16 mA cm<sup>-2</sup>. Similar remark can be made on the *SPCM* at 0.8 mA cm<sup>-2</sup>, whereby owing to low *PC* under this configuration (1.29 kWh m<sup>-3</sup> for 5 h

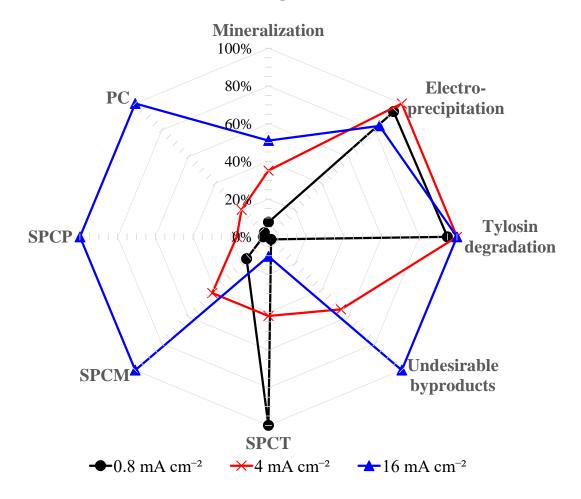
of treatment), it still has margin for longer treatment time to better mineralize organics in the wastewater. Regarding the evaluated *SPCP*, lower *SPCP* was achieved by applying 0.8 mA cm<sup>-2</sup>. It is an inconvenience if the mineral scaling is unwanted, such with the case of EAOP. Electro-precipitation still occurred at about similar degree as 4 mA cm<sup>-2</sup> despite lower current density used. However, this low *SPCP* would represent economical advantage if the electro-deposition on cathode was intended, e.g., in the applications of electro-precipitation for recovery or water softening. Finally, using these values of *SPC*, an estimation on energy consumption to (1) mineralize, (2) degrade pollutant or (3) remove/recover an amount of precipitate could be estimated as function of the duration of treatment performed.

**Table IX.2**. Summary of the evaluated parameters as the outcome of electro-oxidation of RW spiked with 10 mg L<sup>-1</sup> tylosin at 0.8, 4 and 16 mA cm<sup>-2</sup> during 5 h of treatment.

	0.8 mA cm <sup>-2</sup> 4 mA cm <sup>-2</sup>		16 mA cm <sup>-2</sup>	
	(DAC: 1)	(DAC: 5)	(DAC: 20)	
Mineralization (%)	7.7	35.1	51.1	
Electro-precipitation (mg)	53.1	56.6	47.1	
Tylosin degradation (%)	95.0	100	100	
Undesirable byproducts	2.9	82	151	
$(ClO_3$ - $+ClO_4$ - $)$ $(mg-Cl_2 L^{-1})$	2.9	62	131	
PC (kWh m <sup>-3</sup> )	1.29	8.33	41.2	
SPCM (kWh g-TOC <sup>-1</sup> )	0.495	1.27	3.01	
SPCT* (kWh g-Tyl-1)	0.18	0.069	0.017	
SPCP (kWh g-precipitate-1)	0.012	0.074	0.44	

**SPCM**: specific power consumption per gram of TOC removed, **SPCT**: specific power consumption per gram of tylosin degraded (\*: the time depends on the duration of total tylosin removal) and **SPCP**: specific energy consumption per gram of electro-precipitate formed.

From the values of performance provided in **Table IX.2**, **Fig. IX.2** is plotted to visualize relative distribution of parametric performances as function of different  $j_{app}$ . For that, the evaluated value at a given  $j_{app}$ , was expressed relatively to the highest one and it was done for all the inspected parameters. Thus, according to **Fig. IX.2**, it was clearly observed that 16 mA cm<sup>-2</sup> outnumbered the other  $j_{app}$  in terms of energetic consumption (*PC*, *SPCM* and *SPCP*) due to high current applied. *SPCT* was the only exception, owing to very fast organic removal. The formation of undesirable byproducts occurred the most using this  $j_{app}$  as described previously. Highest organic mineralization was also obtained at 16 mA cm<sup>-2</sup>, while slightly lower electro-precipitation occurred due to vigorous H<sub>2</sub> evolution. Contrastingly, minimal energy requirement was achieved using low  $j_{app}$ , but at the price of low organics and target pollutant removal. As a result, the *SPCT* was apparently the highest due to much slower kinetics of degradation.



**Fig. IX.2**. Relative performances of the electro-oxidation of reclaimed wastewater spiked with 10 mg L<sup>-1</sup> tylosin at 0.8, 4 and 16 mA cm<sup>-2</sup> during 5 h of treatment.

### IX.6. Porous cathode, the suitable candidate for electro-precipitation recovery

The kinetics of precipitation taking place on non-porous stainless steel was compared to that occurring on carbonaceous graphite plate and carbon paper. They all possessed distinct electroactive surface and porosity. It was demonstrated that the use of highly porous cathode such as carbon paper would only lead to a quickest CaCO<sub>3</sub> precipitation, a highest formation of toxic ClO<sub>3</sub><sup>-</sup> and ClO<sub>4</sub><sup>-</sup> byproducts as well as the necessity of post-treatment. Similarly, when graphite was employed as cathode, more mineral scaling was formed compared to the stainless steel, with the latter being less porous. The results suggested that stainless steel was the most suitable candidate to be used as cathode to minimize the electro-precipitation. Meanwhile, carbon paper and other porous materials would suit the purpose of the electro-precipitation for recovery process as the precipitation was more favorable when porous material acted as cathode.

### **IX.7.** Perspectives

#### **Short terms**

Drastic trend of deposition as well as ohmic drop were observed between 50 and 100  $\mu$ m  $d_{elec}$ . More experiments are needed for the distances in between for a better understanding on the involved mechanism. Improvement could still be made on the theoretical model involving  $Mg^{2+}$  because the experimental trend of  $Mg(OH)_2$  as function of  $d_{elec}$  was not so apparent. In reality,  $Mg(OH)_2$  is more amorphous relatively to  $CaCO_3$ , thus its stability was less reproducible. Furthermore, low  $Mg^{2+}$  concentration (5 mg  $L^{-1}$ ) relatively to  $Ca^{2+}$  (150 mg  $L^{-1}$ ) might complicate the result repeatability. Furthermore, the model could call for additional parameters coming from the contribution of anode especially from the provision of  $O_2$ . This provision should as well vary between interelectrode gaps. The influence of higher concentration of  $Mg^{2+}$ ,  $Ca^{2+}$ , carbonates and other ions such as  $SO_4^{2-}$  could be further investigated. It has been understood from literature that the variation of each can influence the kinetics of precipitation of others. Hence, the robustness of models could be then tested correspondingly.

The experiments were conducted in batch recirculation mode throughout this work, which is a common practice during parametric studies. As a next step, the experiments shall be tested in continuous mode which reflects more on real case scenarii. The validity of developed model could also be tested. Additionally, a correlation to estimate the mass of precipitate deposited as a function of process duration and reactor geometry could be proposed.

A dimensionless *Sh* correlation was established in parallel-plate electrochemical reactor under laminar regime. Its application works best under similar reactor geometry but would require some corrections if it is to be applied for reactors with other geometries. Meanwhile, the mass transfer correlation under turbulent regime in microfluidic reactor has not yet been done. It would be an interesting contribution to the literature.

#### **Medium terms**

A program involving the overall modeling should be tempted and could not only consider the electroprecipitation but also the pollutants degradation together with the evolution of byproducts in real effluent at a given operating conditions. Besides, only inorganic byproducts were monitored during the EO process. The fate of organics, including organohalogen byproducts and any possible volatile organic compounds could also be quantified in order to get a complete mass balance. So, a complete comparison with the emission regulations could be made. In addition, the tylosin parent molecule was efficiently degraded but its byproducts were not identified. The intermediates could be more toxic than the pristine compound. Therefore, it might be of particular interest to obtain a degradation pathway of tylosin by EAOP, since it has not yet been documented in literature. Moreover, the study of the removal of tylosin micropollutant (instead of spiked concentration) at a concentration typically detected in manure wastestream (in the order of µg L<sup>-1</sup>) seems more practical. It was the initial objective of the thesis but could not be completed due to time constraints. To achieve the purpose, the use of powerful analytical tools, e.g., high performance liquid chromatography with tandem mass spectroscopy (LC-MS/MS) to follow its degradation and evolution of byproducts shall be required.

#### Long terms

After series of experiments with various  $j_{app}$  by treating synthetic and real wastewater, it has been shown that  $j_{app}$  in the region of 1 to 2 mA cm<sup>-2</sup> might present a good compromise considering the overall power consumption, occurrence of mineral scaling, total organic mineralization, pharmaceuticals degradation as well as the evolution of undesirable byproducts. However, the outcome of this thesis suggested that, by drastically changing scale from millimetric to microfluidic environment, not only the mechanism of electro-precipitation remained the same, but its formation was also favored by less  $H_2$  evolution activity and more favorable  $OH^-$  production on cathode surface. As a result, more deposition was generally observed under submillimetric configuration. At this stage, microfluidic feature might not be able to challenge the status quo of polarity reversal technique applied for descaling purpose as the quest to find its alternative shall continue.

In the meanwhile, a particular setup was found during the thesis work that could significantly slow down the kinetics of electro-precipitation. It involved the use of graphite as cathode with 100 µm interelectrode gap, a setup by which the formation of OH was minimized. Though, adopting such configuration seemed impractical to treat high volumes of effluent in real field water/wastewater treatment application, operating several microfluidic cells in parallel to improve productivity merits a trial. Total treatment duration ought to be longer to maximize organic removal but it would not cause energetic concern when submillimetric reactors are adopted. Furthermore, the deposits formed after 5-h of EO treatment did not result in noticeable decrease in charge transfer and organic degradation efficacy. Thus, it indicated that the electro-oxidation efficiency was not yet impacted by the cathodic scaling. This upscaling process shall be operated in continuous mode and at higher flow rates to get closer to the real field practice.