

Multiphase reactors modelling & process design for environment friendly production of aluminium



within EU H2020 ALSiCal Project



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*PhD thesis supervised by:
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This project has received funds from the European Union's Horizon 2020 research and innovation programme under Grant Agreement No 820911

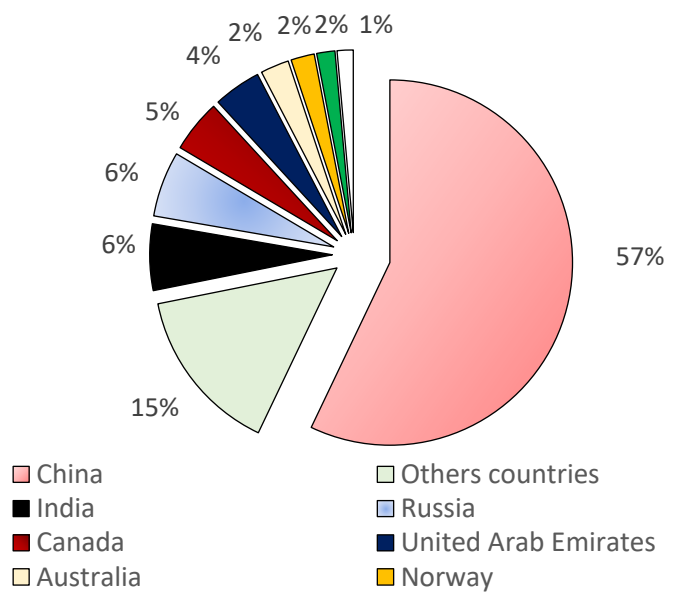
CONTEXT: Aluminium production

Key figures [1]:

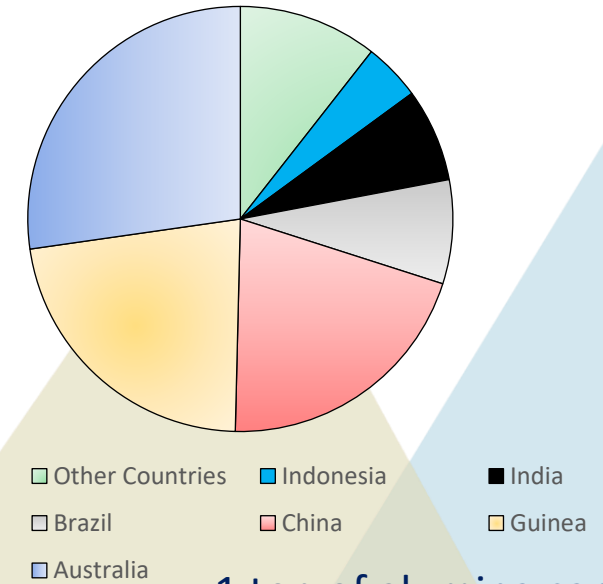
- Worldwide aluminium consumption in 2018: 63 millions tons
- Traditional process: Bayer Patent (1888) using bauxite as raw material
- Consumption of bauxite for 1 ton of alumina varies between 1.9 to 3.6 tons
- 370 millions tons of bauxite mined in 2019



Use of Aluminium in the world in 2018



Bauxite Mine Production in the world in 2019



1 ton of alumina can produce 2 tons of **red mud** (140 millions tons per year), an hazardous waste for the environment (high pH, fine particles, heavy metals, etc...)



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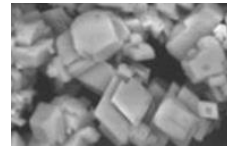
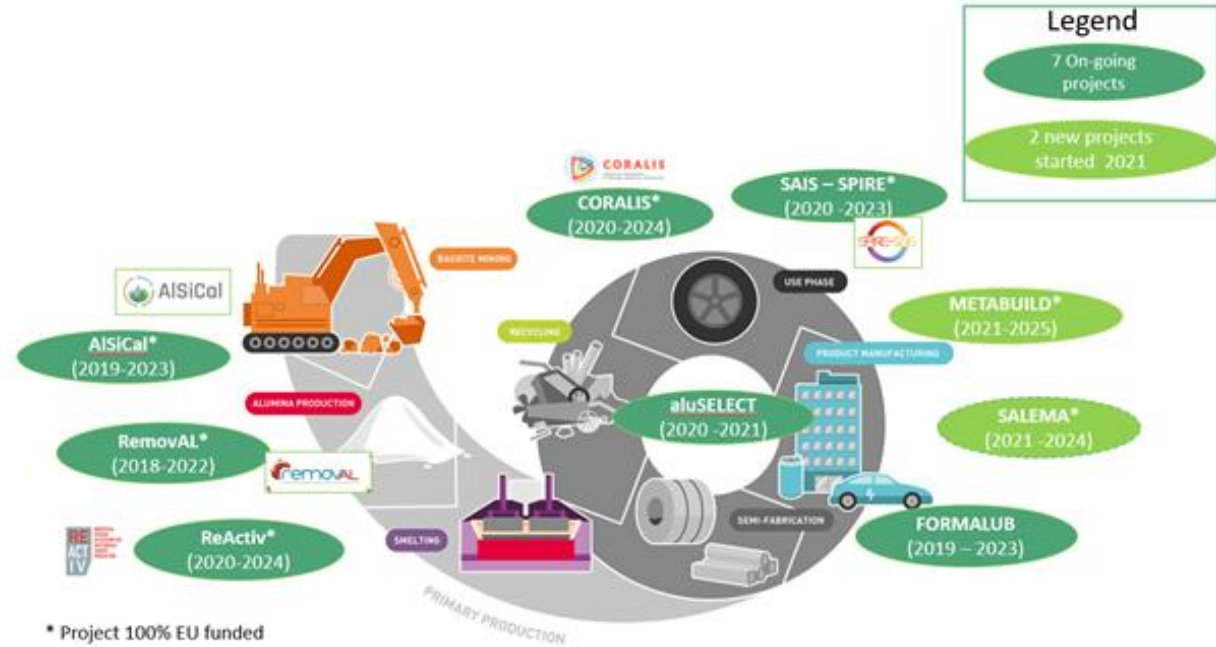
[1] USGS

CONTEXT: AlSiCal project

EU concern for sustainable aluminium production

AlSiCal project (lead by IFE, Norway) which aims to produce « green » aluminium

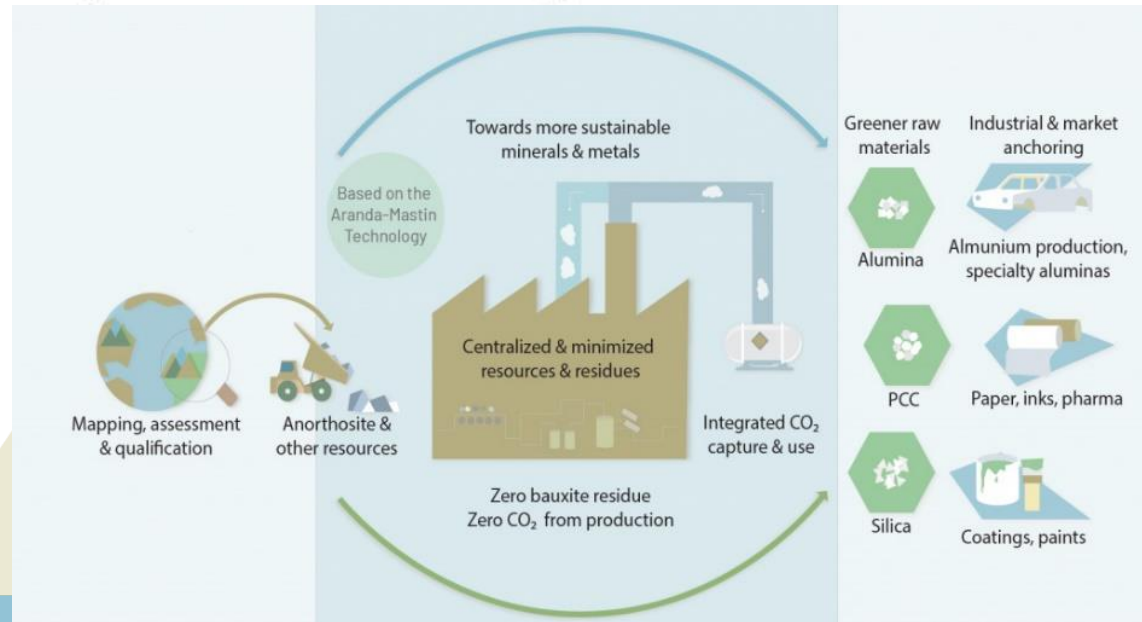
- from **anorthosite**, an abundant feldspar found for instance in Greenland, Norway & Canada
- with **integrated CO₂ utilization**
- with **valorisation of each by-product** (silica, precipitated calcium carbonate)



16 partners, from 9 different countries



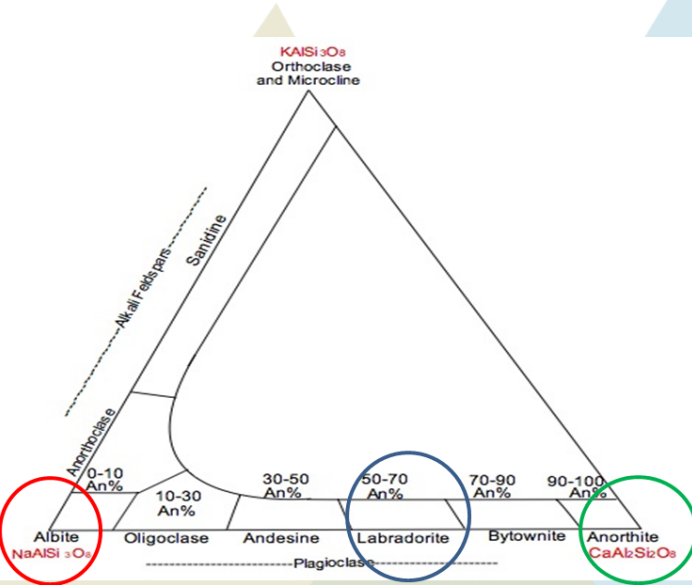
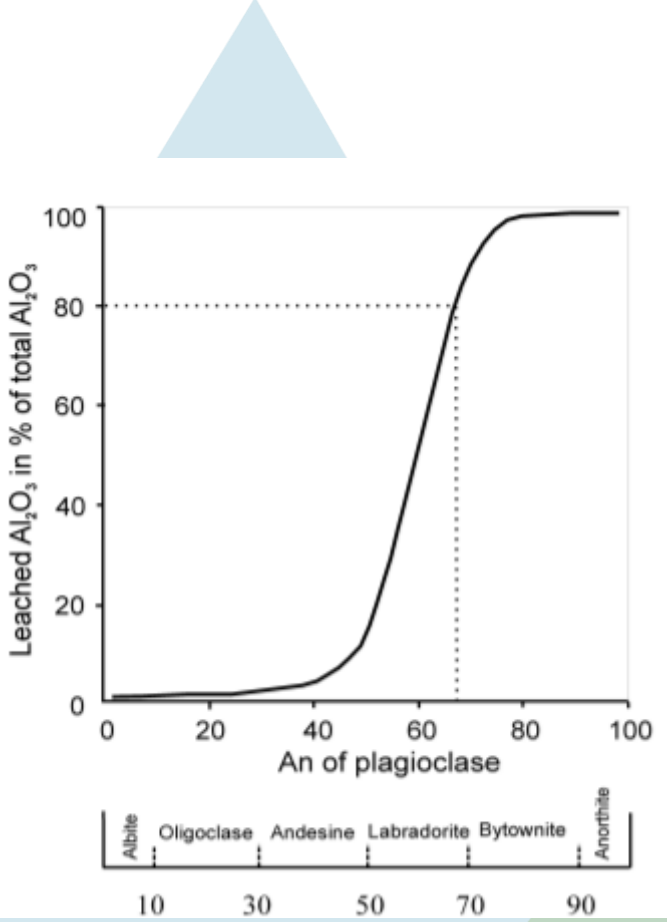
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CONTEXT: Aluminium production

Anorthosite

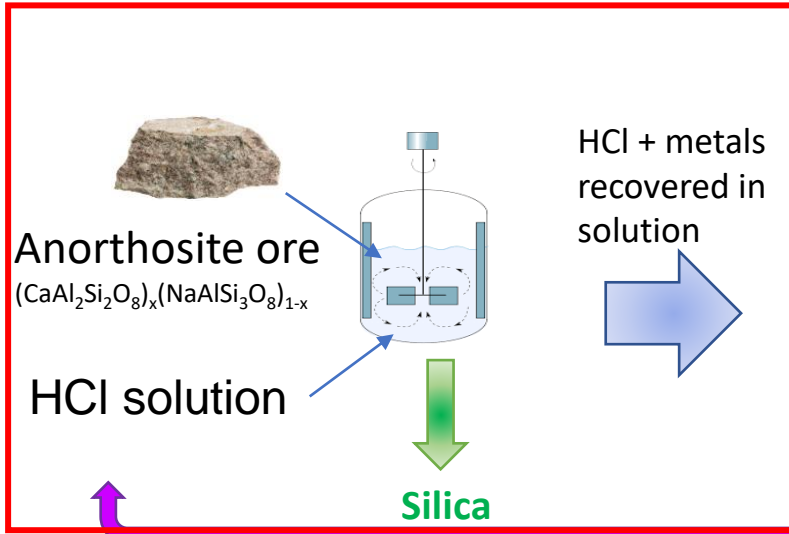
- Plutonic rock
- Composed of plagioclase feldspar (over 90% of the rock is composed of this mineral)
- Plagioclase is a very common mineral group, more than half of the Earth's crust is composed of it
- $(CaAl_2Si_2O_8)_x(NaAlSi_3O_8)_{1-x}$
- The more the % of Anorthite ($CaAl_2Si_2O_8$), the easier is the leaching: x should be > 0.5



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My PhD thesis in AlSiCal project

to develop phenomenological reactor models for process optimization & scale-up

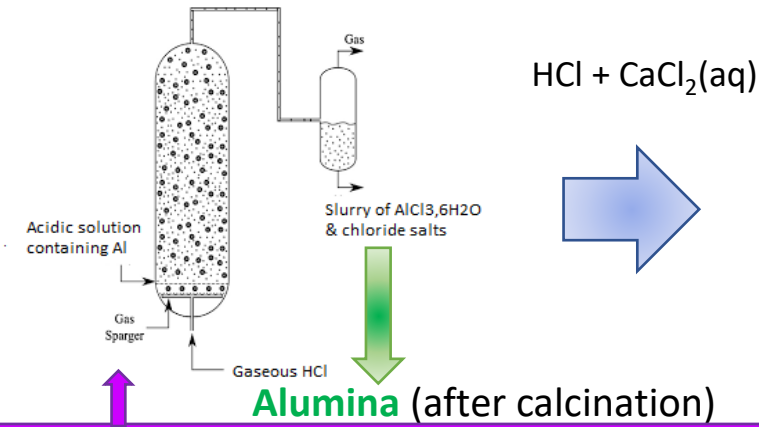


Leaching of the anorthosite

To **provide recommendations** on the operating conditions for the **dissolution of anorthosite** into desired products & to **predict the experimental extraction yields** with respect to process inputs (e.g. particle size distribution)



H^+ activity is a key parameter of ore dissolution rate
Many silicate solids could be formed at too low [HCl]

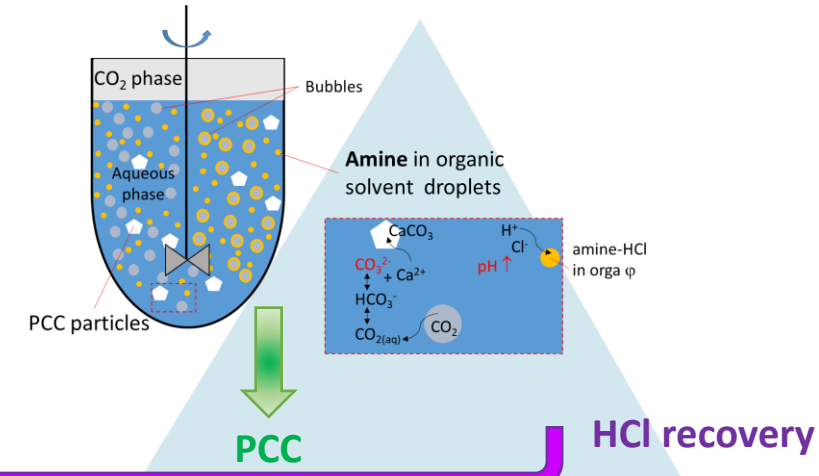


Sparging & crystallization reactor

To **guide the operating conditions** and the number of operation steps to optimize both **aluminium recovery** and **product purity**
→ includes reactive gas absorption, crystal nucleation & growth



High value of ionic force requiring to use Pitzer model for activity coefficient calculation



L-L extraction & carbonation reactor

To develop a model which takes into account **3 simultaneous phenomena**:

- The **precipitation** of the calcite
- The **dissolution** of the gas (CO_2), potentially enhanced by the organic phase (amine) and chemical reactions
- The **reactive extraction** of HCl by the amine, impacting the precipitation of the calcite (by increasing the pH)



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Selection of activity coefficient model, Predicted solid phase speciation (comparison of geochemical databases), Selection of an operating window

Kinetic modelling

Dissolution model hypotheses, Critical analysis of literature kinetics & optimization of rate parameters, Parametric study, Comparison with available experimental data (patent)

Evaluation of reactor configurations toward a continuous process

Relevant geochemical tools

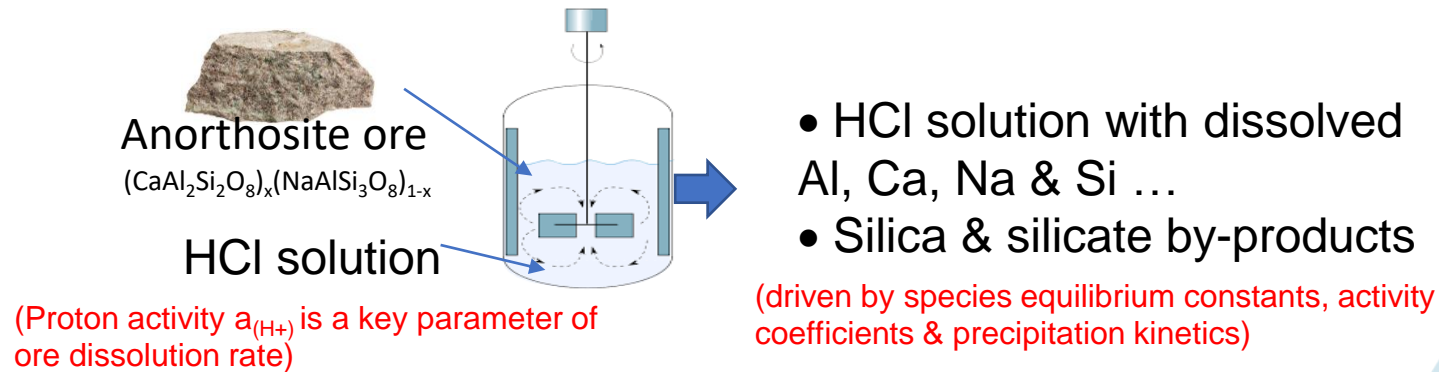
Modelling of leaching reactor: overall strategy

My PhD thesis in ALSiCal project

CONTEXT: Aluminium production



Modelling of leaching reactor: overall strategy



Objectives:

- give a first-hand prediction of the product speciation in the aqueous and solid phases
- provide the means to investigate the sensitivity of the process to key operating variables, including HCl & ore concentrations, feed particle size distribution, temperature ...
- predict process performance to select favourable operating conditions
- provide guidance for reactor scale-up



Modelling of leaching reactor: overall strategy

Equilibrium calculations (PhreeqC, OLI Syst. ...)

- to help define the likely operating window for high extraction yield,
- to anticipate the formation of solid by-products (other aluminosilicates),

Coupling of thermodynamics & dissolution kinetics

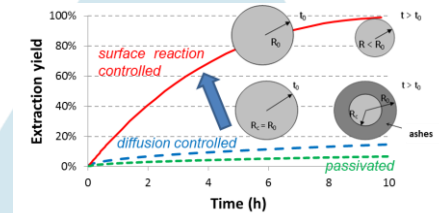
- to predict the time-yield curves with respect to main process inputs

Validity and completeness of thermodynamic databases

Experimentations



- Dissolution tests
- Liquid analysis & solid product characterization



Identification of actual rate-limiting elementary process(es)

Optimization of rate parameters (ex: k_0 , E_A ...)

Reactor model embedded in a simulation / optimization environment

Challenges:

- electrolyte solutions with ionic strength up to 10 mol/kg → selection of suitable activity coefficients
- no information on possible solid by-products → comparison of predictions from several databases
- dissolution yield depending on the active surface → PSD of the ore to be accounted for
- variability of the reactivity of anorthosite & delay on experimental data due the Covid situation

→ model assessment & optimization from relevant literature data



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Relevant geochemical tools: PHREEQC vs. OLI



Open source geochemical code
developed by USGS

- Open source code & possibility to modify/extend databases
- PhreePlot software for advanced graphs
- Possibility to implement shrinking core/particle models
- Can be easily coupled with Matlab, Python, Comsol, R languages/softwarewares ... for advanced calculations

Advantages



Commercial thermodynamic software

- Activity coef. models for very high ionic strength (esp. Mixed Solvent Electrolyte model)
- User-friendly interface
- Team of thermodynamic experts available

Disadvantages

- Limitations of standard B-dot model (low I range), limited Pitzer database

- Database extension/modification as an option
- Standard kinetic models not suitable
- No available coupling with Matlab (extra developments needed for using Oli Eng. Dev.)

Preferred tool for kinetic modelling

Used as thermodynamic benchmark



Relevant geochemical tools: PHREEQC vs. OLI



ALSiCal



DATABASES

(Na, Ca, Al, Si, Cl elements)



Thermoddem 2017:

128 solids, 40 aqueous species

Solids found to precipitate:

- Albite $\text{NaAlSi}_3\text{O}_8$
- Amorphous silica (crystalline SiO_2 deactivated)
- Diaspore $\text{AlO}(\text{OH})$
- Prehnite $\text{Ca}_2\text{Al}_2\text{Si}_3\text{O}_{10}(\text{OH})_2$
- Kaolinite $\text{Al}_2\text{Si}_2\text{O}_5(\text{OH})_4$
- Pyrophyllite $\text{Al}_2\text{Si}_4\text{O}_{10}(\text{OH})_2$
- Laumontite $\text{Ca}(\text{Al}_2\text{Si}_4)\text{O}_{12} \cdot 4\text{H}_2\text{O}$
- ~~Heulandite(Ca) $\text{Ca}_{1.07}\text{Al}_{2.14}\text{Si}_{6.86}\text{O}_{18} \cdot 6.17\text{H}_2\text{O}$~~
- ~~Beidellite(Na) $\text{Na}_{0.34}\text{Al}_{2.34}\text{Si}_{3.66}\text{O}_{10}(\text{OH})_2$~~
- ~~Gismondine $\text{Ca}_2\text{Al}_4\text{Si}_4\text{O}_{16} \cdot 9\text{H}_2\text{O}$~~
- ~~Chabazite $\text{Ca}(\text{Al}_2\text{Si}_4)\text{O}_{12} \cdot 6\text{H}_2\text{O}$~~
- ~~Phillipsite(Na) $\text{NaAlSi}_3\text{O}_8 \cdot 3\text{H}_2\text{O}$~~

(only in Thermoddem)

MSE/GEOCHEM (no theoretical limit on I):

79 solids, 16 aqueous species (H_3O^+ instead of H^+ species)

Solids found to precipitate:

- Albite $\text{NaAlSi}_3\text{O}_8$
- Amorphous silica (crystalline SiO_2 deactivated)
- Boehmite $\text{AlO}(\text{OH})$
- $\text{Al}(\text{OH})_3$
- Prehnite $\text{Ca}_2\text{Al}_2\text{Si}_3\text{O}_{10}(\text{OH})_2$
- Kaolinite $\text{Al}_2\text{Si}_2\text{O}_5(\text{OH})_4$
- Pyrophyllite $\text{Al}_2\text{Si}_4\text{O}_{10}(\text{OH})_2$

AQ/GEOCHEM ($I \leq 30 \text{ mol/kg}_w$)

58 solids, 20 aqueous species

Same solids found to precipitate as with MSE/GEOCHEM:

+ Laumontite $\text{Ca}(\text{Al}_2\text{Si}_4)\text{O}_{12} \cdot 4\text{H}_2\text{O}$

+ AlOHCl_2 (not found in Thermoddem)

Selected for 1st step
(H^+ , $I < 15 \text{ mol/kg}_w$)

→ Reduction of Thermoddem database to compare solid phase speciation



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Thermodynamics: Selection of activity coefficient model

$$\text{BDOT: } \log_{10} \gamma_i = -\frac{A_{\text{DH}} z_i^2 \sqrt{I}}{1 + B_{\text{DH}} a_i \sqrt{I}} + bI \quad (I \leq 1 \text{ M})$$

$$\text{PITZER: } \ln \gamma_i = \frac{1}{RT} \left(\frac{\partial G^E}{\partial n_i} \right)_{T,P,n_s,n_j}$$

$$\frac{G^E}{RTw_s} = f(I) + \sum_i \sum_j \lambda_{(I)}(i;j) m_i m_j + \sum_i \sum_j \sum_k \mu(i;j;k) m_i m_j m_k$$

$$\text{SIT: } \log_{10} \gamma_i = -\frac{A_{\text{DH}} z_i^2 \sqrt{I}}{1 + B_{\text{DH}} a_i \sqrt{I}} + \sum_k \varepsilon_{(T)}(i;k) m_k$$

$$\text{OLI/AQ: } \log_{10} \gamma_{\pm} = -\frac{A|Z_+Z_-|\sqrt{I}}{1 + \sqrt{I}} + \frac{(0.06 + 0.6B)|Z_+Z_-|\sqrt{I}}{\left(1 + \frac{1.5}{|Z_+Z_-|} I\right)^2}$$

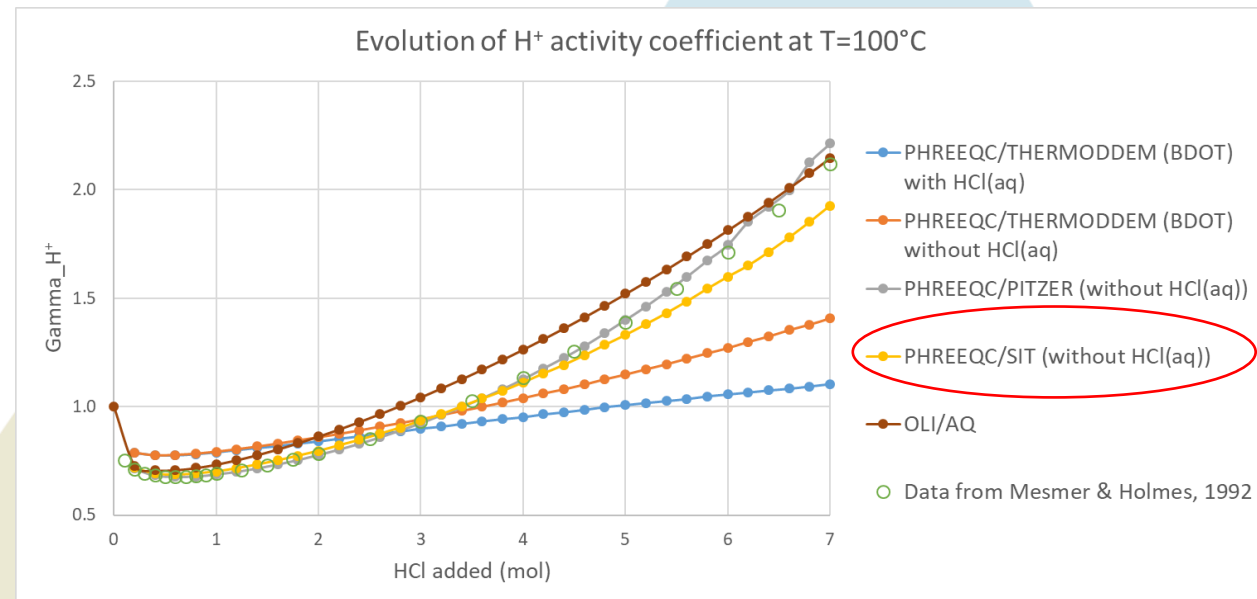
(Bromley-Zemaitis)

$$+ BI + CI^2 + DI^3$$

→ OLI/AQ predicts almost complete dissociation of HCl:
HCl(aq) can be omitted

→ PHREEQC/SIT gives similar γ_{H^+} values as PHREEQC/PITZER
or OLI/AQ

➤ Predictions for the solvent (HCl)



I up to 8 mol/kg_w

→ SIT to be evaluated for acidic dissolution of anorthosite



➤ Application to acidic anorthosite dissolution

Construction of a new database

SIT database:

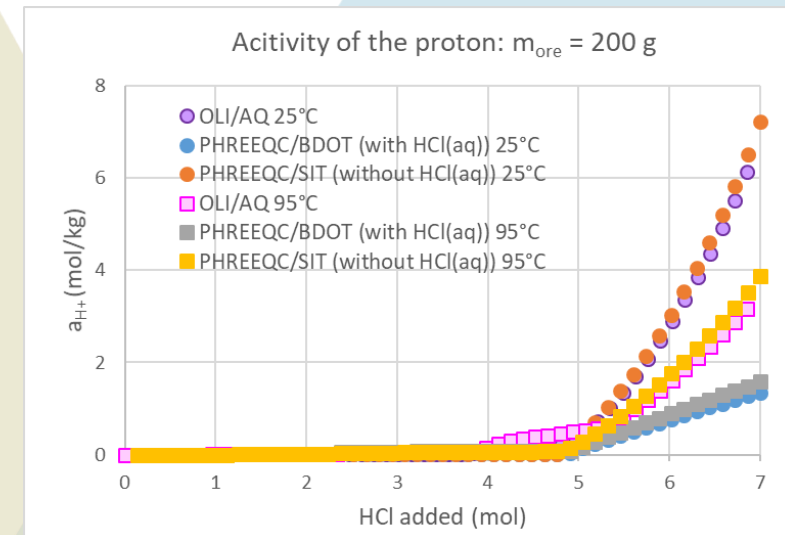
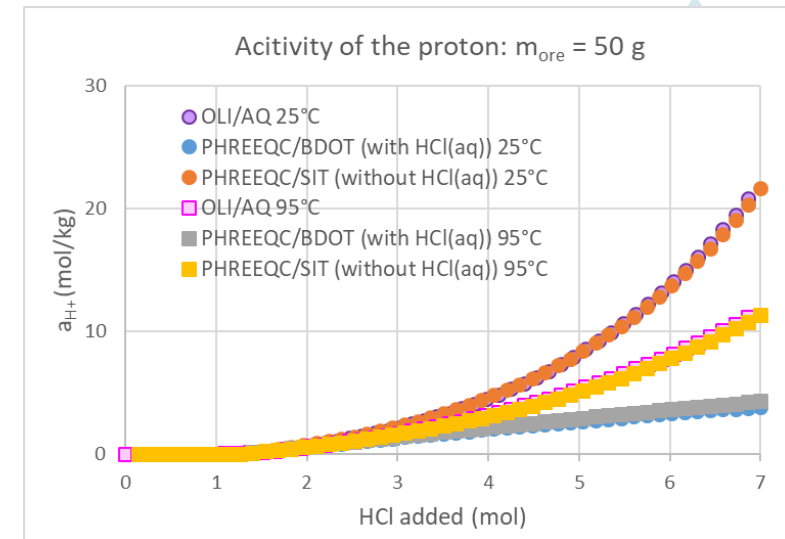
$\varepsilon(\text{H}^+; \text{Cl}^-)$, $\varepsilon(\text{Ca}^{2+}; \text{Cl}^-)$, $\varepsilon(\text{Na}^+; \text{Cl}^-)$, $\varepsilon(\text{Al}^{3+}; \text{Cl}^-)$ } at 25°C
 $\varepsilon(\text{Na}^+; \text{H}_3(\text{SiO}_4)^-)$, $\varepsilon(\text{Na}^+; \text{H}_2(\text{SiO}_4)^{2-})$...

+ $\varepsilon(\text{H}^+; \text{Cl}^-)$, $\varepsilon(\text{Ca}^{2+}; \text{Cl}^-)$, $\varepsilon(\text{Na}^+; \text{Cl}^-) = f(T)$ (Xiong, 2006)

+ mineral phases from Thermoddem 2017

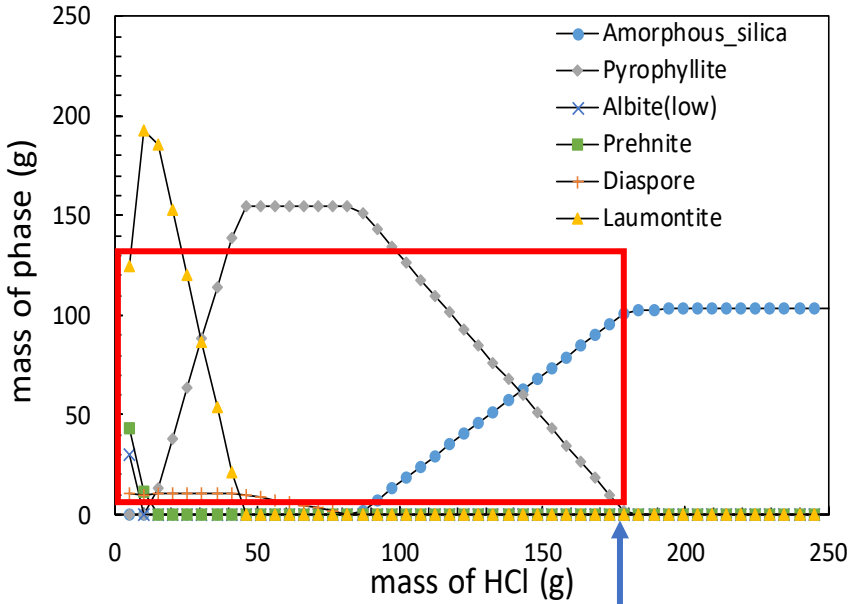
→ Higher a_{H^+} for SIT vs. B-dot

→ Predictions of SIT in agreement with OLI/AQ

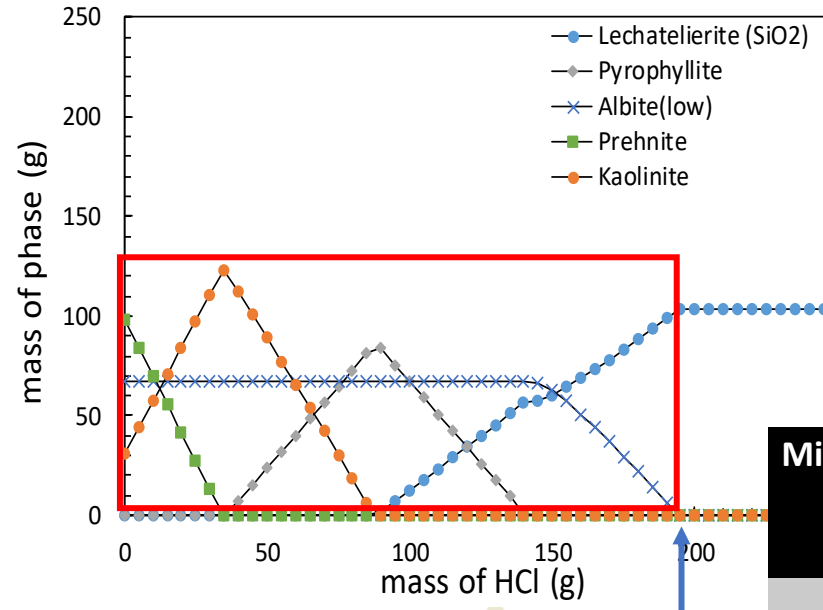


Thermodynamics: Predicted solid phase speciation

(a) PHREEQC/SIT



(b) OLI/AQ



200 g of $Ca_{0.65}Na_{0.35}$ anorthosite at 95°C as a function of HCl amount added to 1 kg of water

- Total amount of HCl required to dissolve all Al-bearing minerals is rather consistent for the 2 databases
- Discrepancies in solid speciation at lower [HCl] can be explained by slightly different thermodynamic data

Mineral name	Difference in Gibbs energy of formation ΔG_f° @25°C (kJ/mol) (Thermoddem – OLI/GE)
Anorthite	-15.5
Kaolinite	-4.9
Albite(low)	50.1
Prehnite	-4.7
Pyrophyllite	-11.1
Laumontite	-16.4
Amorphous_silica /Lechatelierite	-1.4
Diaspore	-9.0

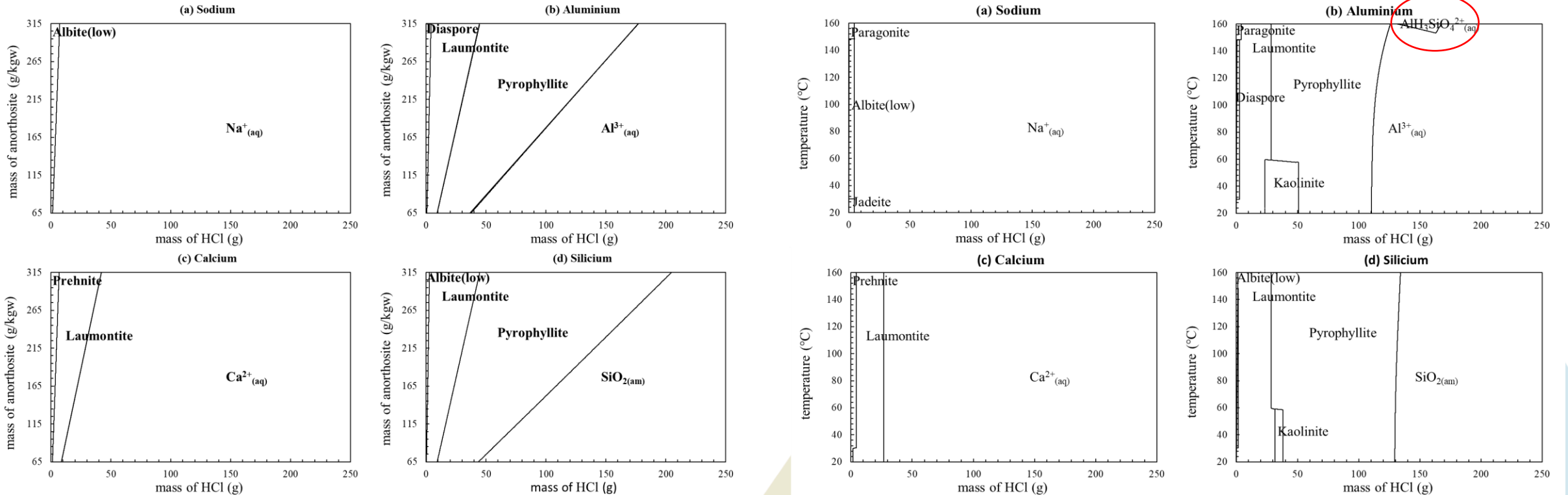


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(values of ΔG_f° between -850 & -6700 kJ/mol)

Thermodynamics: Selection of an operating window

➤ Influence of parameters (T, [HCl], S/L, ...)



T=95°C

m_{ore} = 200 g

- The higher is the initial mass of ore, the more HCl is needed
- Rather low effect of T, but complexation of silicon in solution above 150°C ($\text{Al}^{3+} \rightarrow \text{AlH}_3\text{SiO}_4^{2+}$)
- 20 wt.% HCl solution can achieve complete Al, Ca & Na dissolution for S/L as high as 25 wt.%



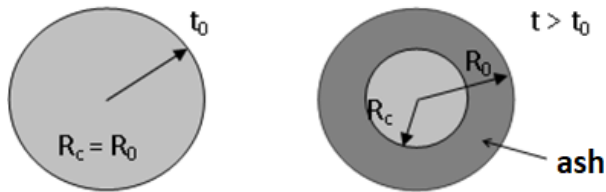
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Kinetics: construction of the acidic dissolution model

- Choice of a dissolution model
- Critical analysis of literature kinetic data → Selection of dissolution rate model & relevant experimental data (Gudvangen anorthosite)
- Optimization of rate parameters
- Parametric study to finalize the choice of operating conditions: T, PSD ... and comparison with an experimental point

Dissolution kinetics

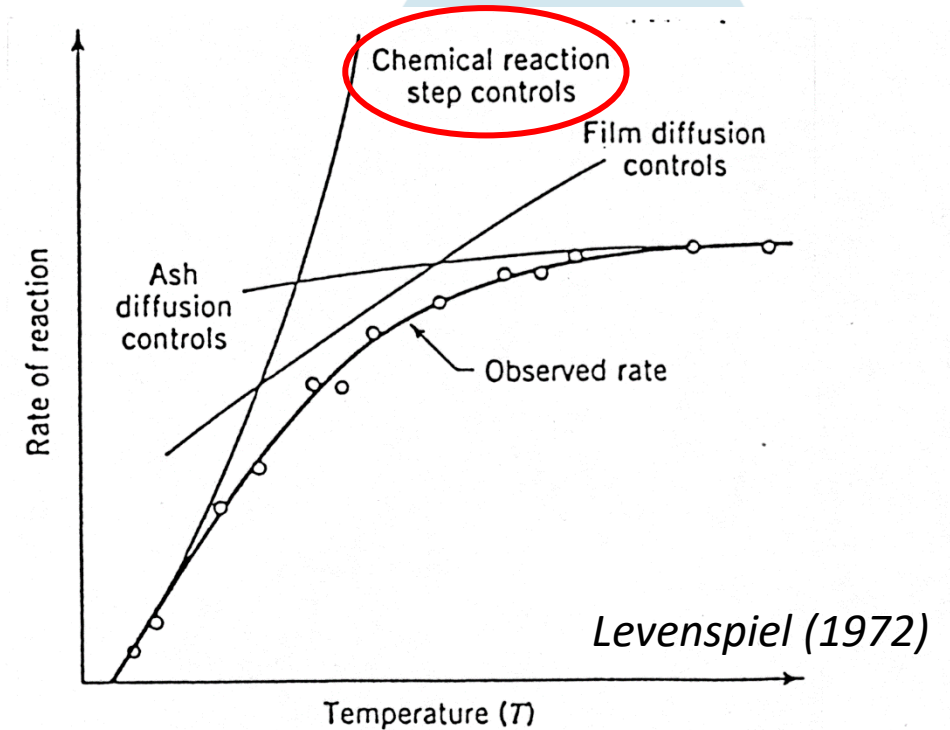
based on shrinking core model with surface reaction control
(no limitation by diffusion in the ash layer)



$$\frac{dn_i}{dt} = -r \cdot SSA_{0i} \cdot w_{0i} \cdot m_0 \left(\frac{n_i(t)}{n_{i0}} \right)^{2/3}$$

$$r = k_0 \cdot \exp(-E_a/RT) \cdot a_{(H^+)}^n \cdot a_{(Al^{3+})}^m \cdot (1 - 10^{SI})$$

SSA_{0i} : specific surface area of class i, m_0 : ore mass
 w_{0i} / n_{0i} : initial weight fraction / number of moles of class i



Kinetics: construction of the acidic dissolution model

Matlab code to couple optimization toolbox with the kinetic model embedded in Phreeqc for parameter fitting



- Script divided in three parts to isolate unchanged blocks & « kinetic block » with reaction rate parameters to be optimized

- « **Kinetic block** » : time-integration of the mass balances & calculation of the speciation (solids & solution) at each step for **given reaction rate parameters**

1st module

2nd module



MATLAB

- Reading of experimental data
- Initial parameter set
- **Least squares optimization: *fminsearch* function** (Nelder-Mead simplex method)
- Optimal parameter set
- **Final comparison** between experimental data & model predictions



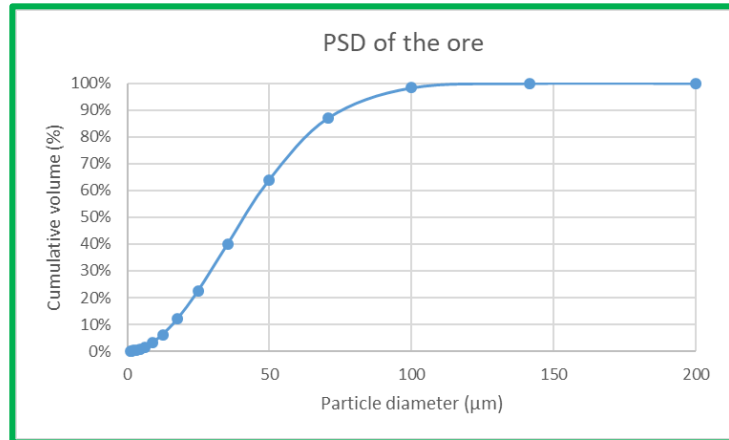
Kinetics: Optimization of rate parameters

- ANORTAL project data (Gjelsvik, 1980)

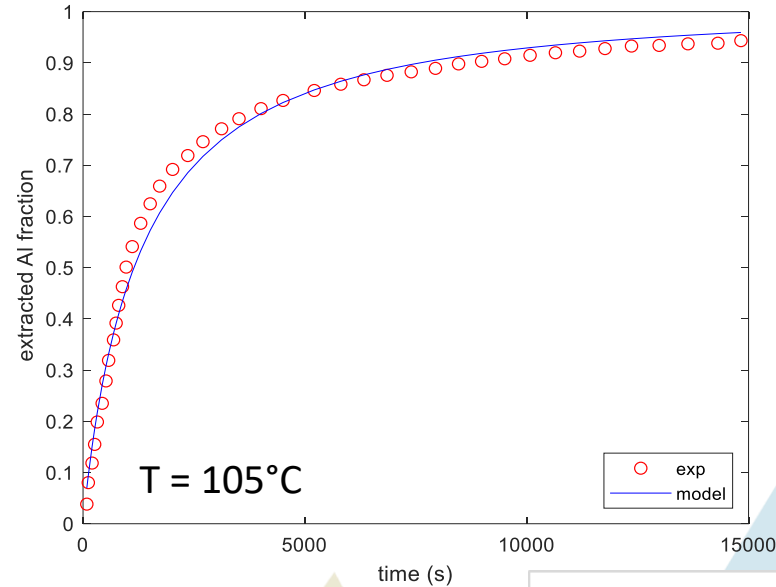
Description of PSD

16 classes (SSA_{0i} , w_{0i})
generated from Rosin Rammler model
(α & d_{90} parameters)

Passing fraction = $1 - \exp(-(d_p/d^*)^\alpha)$



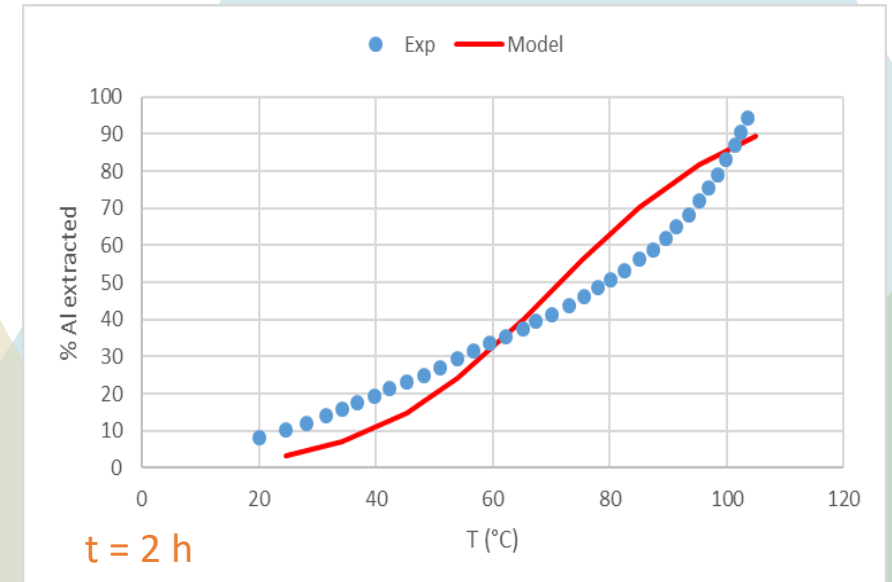
(d_{90} of $75 \mu\text{m}$ with $\alpha = 2$)



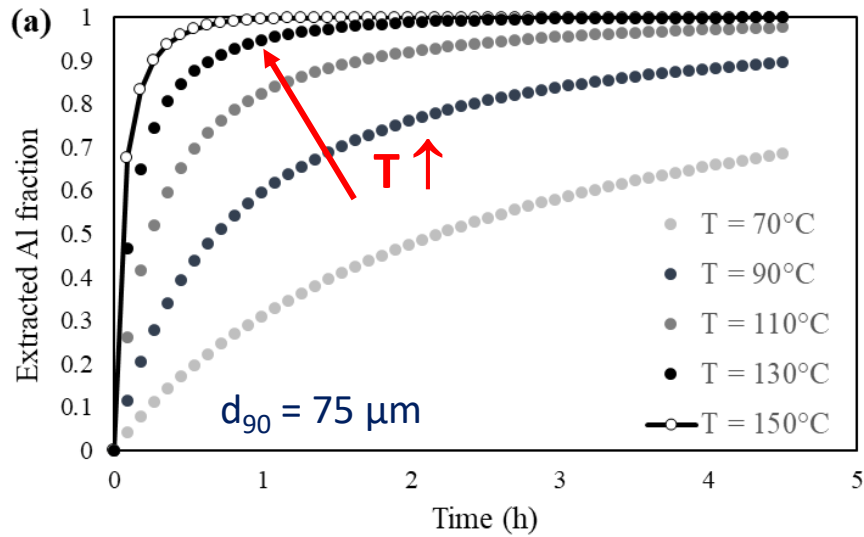
Gudvangen anorthosite
S/L = 23 wt.%, 6N HCl solution

Optimized values:

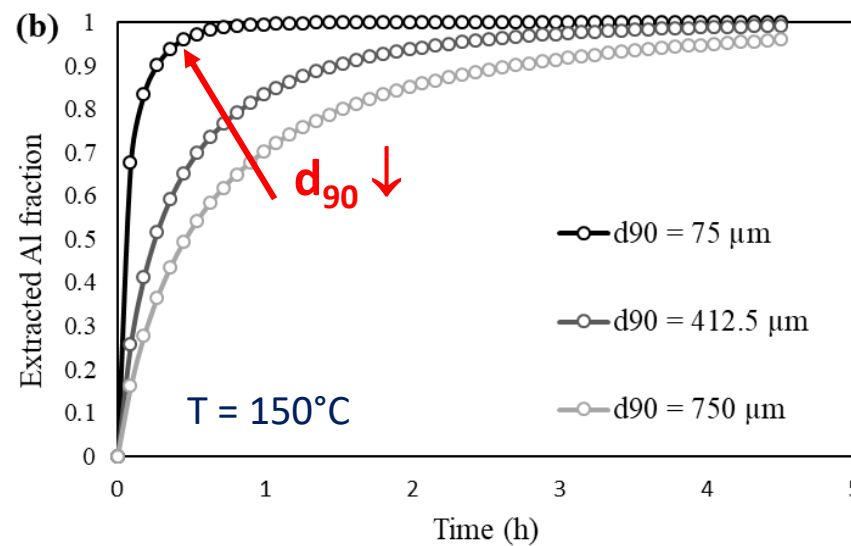
$E_A = 65.9 \text{ kJ/mol}$,
 $k_{105^\circ\text{C}} = 1.03 \cdot 10^{-9} \text{ mol.cm}^{-2}.\text{s}^{-1}$
($n = 0.626$, $m = 0$: set values
following Palandri and
Kharaka, 2004)



Kinetics: Parametric study



$\text{Ca}_{0.65}\text{Na}_{0.35}$ anorthosite treated by 6N HCl
(S/L = 23 wt.%)



As expected :

- the higher the temperature, the faster the leaching
- the smaller the particles, the faster the leaching.



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Kinetics: Comparison to Aranda-Mastin patent data

Aranda & Mastin patent (2015)

100 Liters reactor

10.05 kg of anorthosite

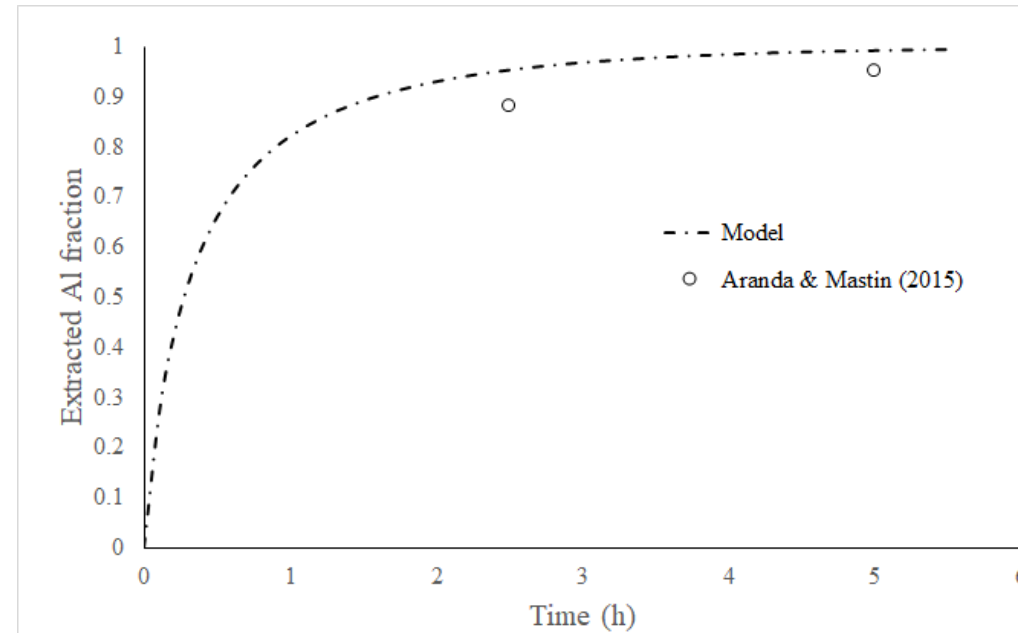
46.75 kg of HCl solution (19.7 wt.% HCl)

T=140°C

Particle size is ~ 300 μm

Extracted Al fraction is :

- 88% at 2.5 h
- 95% at 5 h

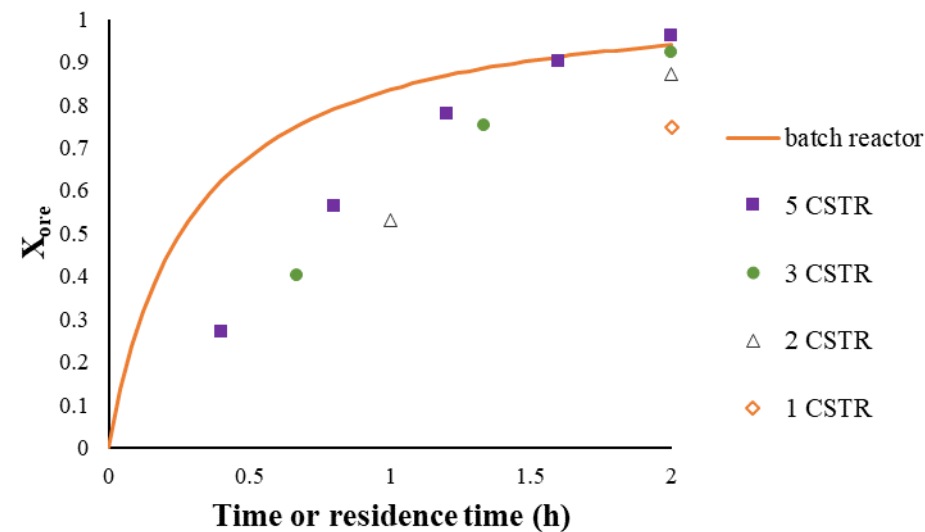
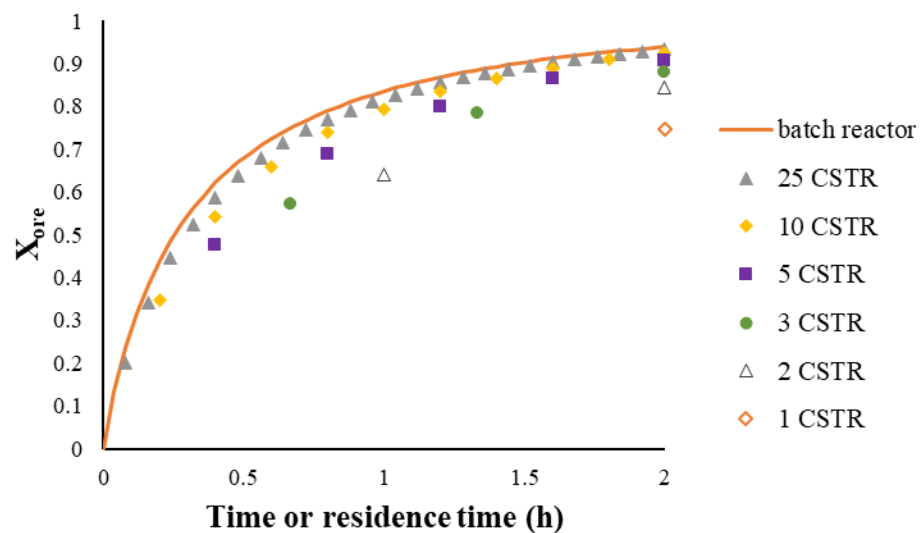
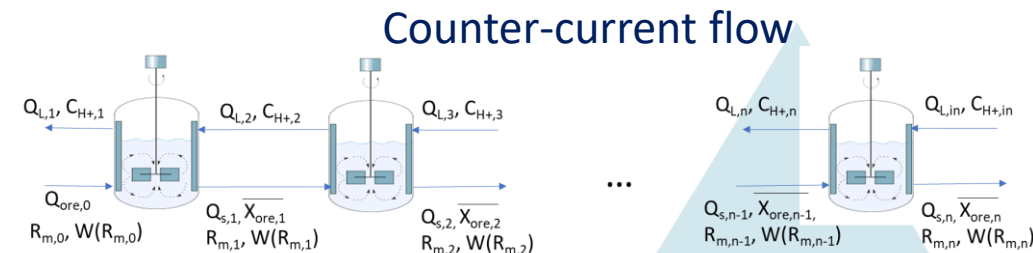
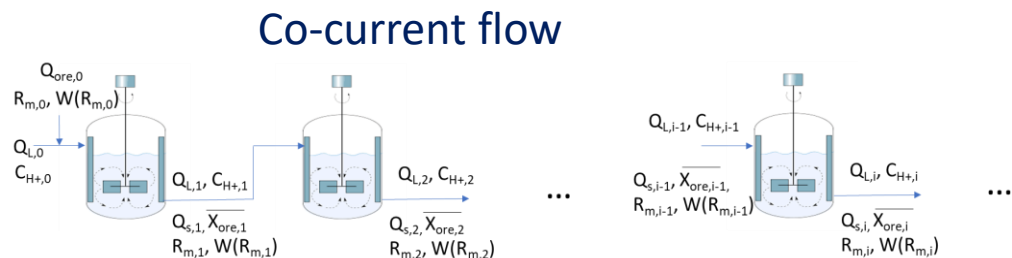


Simulation with polydisperse ore
(d_{90} of 300 μm with $\alpha = 2$)

→ Validation of the dissolution model at higher temperatures



Evaluation of reactor configurations toward a continuous process



Evolution of the conversion extent of anorthosite as a function of the reaction time (batch reactor) or mean residence time (continuous process) for a series a CSTR operating (a) with co-current flow of solid and liquid and (b) with counter-current flow ($T = 150^{\circ}\text{C}$, feed acid concentration = 20 wt.% HCl, $S/L = 23$ wt.%, total residence time = 2 hours, polydisperse ore with $d_{90} = 412.5 \mu\text{m}$).



Conclusions & perspectives

Modelling of leaching reactor:

- A model of batch/continuous stirred reactor (series) is available for Ca-rich anorthosite dissolution
- Model hypotheses (shrinking core under chemical reaction control), thermodynamic environment (selected activity coefficient model & geochemical database) and optimized parameters seem relevant with respect to the Aranda-Mastin patent data
- This model has been further assessed against extraction yields measured by IFE partner in confidential conditions and is able to conveniently predict the observed trends
- An article has been submitted on the theoretical aspects and construction of the leaching model to Minerals Engineering and is currently under review

Perspectives:

Modelling of the two other steps (sparging/crystallization reactor & extraction/carbonation reactor) is under progress

