Nepheline Syenite as an Alternative Source for Aluminum Production

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Abstract: The increasing demand for aluminum in conjunction with the limited available bauxite deposits in Europe results in the dire need for the evaluation of alternative raw materials for the whole value chain of alumina production. The present paper focuses on the possible use of nepheline syenite, originated as a mine byproduct, in alumina production through leaching with an azeotropic HCl solution. Previous work on nepheline syenite dissolution focused more on the extraction of potassium and sodium values. In this work, emphasis is given at the characteristics and leachability of aluminum content phases by applying a high temperature HCl leaching in untreated material. From this point of view, leaching experiments were conducted on nepheline syenite, at a temperature range of 90 °C to 150 °C and with aqueous azeotropic HCl solution. Leaching, in the aforementioned conditions, is a process, characterized by medium aluminum dissolution. The results show that the aluminum-bearing phases that withstand the dissolution process are mainly the ones of sodium and potassium feldspars.

Keywords: nepheline syenite; acid leaching; alternative alumina resources

1. Introduction

Aluminum is a necessary technological material. Its inherent physicochemical properties, such as low density, in conjunction with properties that derive from the formation of alloys make it suitable for numerous applications in the fields of transportation, engineering, construction, building and packaging [1,2]. As a result of these properties, aluminum can be processed into a wide range of consumer goods and, in addition, has a high value as a recycling material, as the re-melting of aluminum scrap requires only 5% of the total energy required for the extraction of the same amount of raw material from ores. Data provided by Sverdrup [2] and Allwood [3] lead, with relative certainty, to two important conclusions. First of all, aluminum is projected to maintain and strengthen its position as a high-tech commodity. Secondly, according to existing data, primary production of aluminum from ores will peak much earlier than the projected peak in demand. In other words, the demand for aluminum should remain high for the next 60 years, but the possibility of producing quantities that will meet this demand with existing sources and technologies is debatable.

In the last 100 years, numerous methods have been developed for the extraction of aluminum from its ores. However, its current commercial production method is based on two energy-intensive processes developed in the late 19th century: the hydrometallurgical Bayer method, which produces pure alumina (Al₂O₃) from bauxite ores, and the electrolytic Hall–Héroult method, which produces pure aluminum metal using as a raw material metallurgical alumina [4]. In the field of alumina production, Bayer process, contributing to 95% of alumina production, is also connected with drawbacks from environmental legislations...
nowadays. More specifically, the environmental impact of the method, particularly the production of the bauxite residue is of great concern.

At the same time in recent years, the availability of bauxite deposits to be processed by the Bayer method in Europe have decreased with consequent increase in the cost of their processing [5]. Bauxite inclusion in the EU’s critical raw materials list for 2020 [6] endorsed this point and made clear the need for the utilization of an alternative resource.

Thus, the scientific interest is focused on the production of alumina from silicate rocks with high aluminum content, such as anorthosite, kaolin, nepheline syenite and feldspar or feldspathoid minerals derived from such rocks. In addition, and following the same trend, interest is also shown to the utilization of secondary raw materials coming from the extractive mining and metallurgical industry, such as slags or mining tailings. Research on this field suggests the direct dissolution of these Al sources in strong inorganic acids, without the need for costly preparation [3]. Chronologically, the first interesting invention in the field of acid extraction was that from Gaudernack [7]. Gaudernack’s invention discloses a process for recovering alumina from aluminous silicate raw materials. A complete acid extraction process using hydrochloric acid solution was presented by Boudreault, Alex and Biasotto [8]. Aluminum ores such as nepheline were used as raw materials in these processes. In nowadays data, Aranda and Mastin’s technology [9], taking into account previous research, suggests a process for alumina and calcium carbonate production from aluminum rich materials with integrated CO$_2$ utilization, comprising: leaching of Al-rich materials in concentrated HCl, separating Al from solution by crystallization of AlCl$_3$·6H$_2$O followed by calcination with HCl recovery.

The application of an acid treatment process is gaining attention due to the overpassing of problems of previous decades, such as the absence of available acid-proof materials and the feasibility of recycling of HCl, and some important advantages, such as the relatively simple conversion from ore to alumina and the higher tolerance to the silica content of ores [9]. Regarding acid recovery, many patents [10–12] consider the use of an organic extraction method, using a variety of amines. The main goal is to extract the free HCl from dilute solutions and then recover the concentrated HCl by stripping the amine.

Nepheline syenite is a plutonic igneous rock consisting mainly of nepheline, sodium, and potassium feldspars. In appearance, as a light-colored and coarse grain-sized material, it shows similarities with granite, without containing any free silica [13,14]. Nepheline syenite main reserves are located in Russia, Norway, Canada and Turkey. Nepheline syenite deposits are also detected as byproducts of some mining projects. An example of such a project is found in Russia where nepheline syenite is a byproduct of apatite mining at Khibiny Mountain in the Kola Peninsula [15].

Nepheline syenite’s high Al$_2$O$_3$, Na$_2$O and K$_2$O content make it a material that can compete with feldspars in glass and ceramic industry. More specifically, Nepheline syenite is used as an active agent for the formation of a glassy phase in ceramics that enhance the physical strength of the final product. As for the glass industry, the high aluminum content of Nepheline syenite acts as a booster in resistance to breaking, as well as thermal endurance and chemical durability. [13]. Moreover, in the last years, some studies suggest the use of nepheline syenite as a potassium source for plants, animal, and human nutrients, particularly in countries such as India where the conventional sources (such as sylvite) are not available [13,16].

In the field of Al production, nepheline syenite is seen as an attractive material due to its high Al content but also challenging due to two major problems: (a) the low dissolution rate of aluminum-bearing phases, and (b) the silica gel formation that is observed in high pulp density and low treatment temperatures [17]. Low Al extraction is connected with the existence of feldspars in nepheline syenite mass that are characterized as almost insoluble phases.

Feldspar’s insolubility drove previous research on the inclusion of a pretreatment step prior to leaching that would transform feldspars to easily dissolved phases. More specifically, two paths had been followed. Jena [13,16] proposed the roasting of nepheline-
syenite with CaCl$_2$ at 900 °C. Danjun [18] suggested sintering with an excess 5–10% of sodium carbonate at 760–880 °C for 1.0–1.5 h. Although such processes would provide high Al extraction, the high energy consumption and CO$_2$ footprint that derive from the thermal treatment step are of great concern.

In this work, a hydrometallurgical treatment of a mining waste characterized as nepheline syenite is presented. Leaching with an azeotropic HCl solution in the temperature range of 90–150 °C was studied in order to recover the maximum amount of aluminum. While, at the same time, an almost pure siliceous residue consisted of precipitated silica, and undissolved Na and K-feldspars were produced. The residue can be used as a raw material in the glass and ceramic industry.

2. Materials and Methods

Nepheline syenite used in this study was obtained from GREENNA Mineral AB in Sweden. This sample was obtained as a byproduct from the utilization of a rare earth elements’ deposit. Nepheline syenite is characterized as a non-hazardous material, nevertheless, the significant volumes of the material produced demand a very large deposition area. Therefore, the main objective of the exploitation of this material is the reduction in land used. The sample was ground to 100%—100 µm. Then, it was dried at 100 °C for 24 h and its chemical composition was determined by wet chemical analysis techniques. The sample was solubilized via fusion method with a mixture of Li$_2$B$_4$O$_7$/KNO$_3$ at 1000 °C for 1 h and then was dissolved in a nitric acid solution. The metals in the solution were measured by Atomic Absorption Spectroscopy (AAS) with the use of PerkinElmer 2100 Atomic Absorption Spectrophotometer. Calcium content was determined with XRF. Mineralogical characterization was made by an X’Pert Pro diffractometer (PANalytical, Malvern, UK) with CuKα radiation. Diffraction patterns were recorded between 2 and 70° 2θ, in 0.02° steps and 1 s per step, while for the qualitative identification of mineralogical phases, DIFFRAC.EVA V5.1 software was used.

For the conventional leaching tests, an Amar G2360 glass autoclave was used, equipped with a 1000 mL capacity jacketed glass vessel and a polytetrafluoroethylene (PTFE) lid. The lid is designed with openings and sockets which allow (a) the recording of pressure through a manometer, (b) the release of the pressure through a release valve, (c) the immersion of a thermocouple and a 6-bladed turbine PTFE stirrer, (d) the drawing of samples and (e) the immersion of electrodes for pH measurement. Mechanical stirrer is connected to a PLC unit which allows for the control of the operating parameters while the heating was controlled by a chiller through the circulation of silicon oil in the external layer of the jacketed vessel. During leaching experiments, the leaching agent was placed in the autoclave and was heated. For the leaching agent, laboratory grade hydrochloric acid (>37%) and deionized water were mixed. When the reactor reached a predetermined temperature, the solid sample was placed inside the reactor signaling the start of the experiment. For experiments at temperatures higher than 100 °C where the reactor operated as an autoclave under saturated vapor pressure, the solid sample was inserted at 90 °C and then was very fast heated to the desired temperature together with the solution. The starting point of the experiment in this case was the moment when the system reached the desired temperature. In kinetic experiments, samples were taken at defined time intervals, and at the end of the experiment, the leachate was separated from the solid residues via vacuum filtration. The final pregnant liquid solution (PLS) and the intermediate samples were analyzed by AAS while the residue was dried, milled and its mineralogical content was analyzed by XRD.

HCl fuming test was performed by adding the azeotropic HCl solution dropwise to the heated solid sample, placed in an evaporating dish, at 110 °C under continuous blending of solids. The droplets partially reacted with the solids and partially were vaporized, leaving back a dried reacted solid material which subsequently was leached with water at ambient temperature for 1 h. The solution obtained was analysed with the same techniques described above. A flowsheet of the conventional leaching and fuming technique tested in this study is depicted in Figure 1.
In this work, the effect of temperature (90–150 °C), pulp density (2.5–10% w/v) and retention time (0–24 h) on the extraction of aluminum was studied during the leaching of the mining waste with an azeotropic HCl solution (5.89 M) under constant stirring rate of 300 rpm. In addition, the co-dissolution of Si, K and Na was monitored in all experiments. Due to silica gelation problems observed at high pulp densities, an HCl fuming test was performed instead of conventional leaching tests to investigate the avoidance of silica gel formation at higher pulp densities.

3. Results

In this section, the chemical and mineralogical characterizations of nepheline syenite are presented in Table 1 and Figure 2. Furthermore, the results from the leaching experiments are presented, giving emphasis to the mineralogical characterization of residues after leaching as a tool to analyze and understand what happens during the leaching process.

Table 1. Chemical analysis of nepheline syenite sample used in this study.

| Oxide   | Mass (wt.%) |
|---------|-------------|
| Al₂O₃   | 22.20       |
| SiO₂    | 54.80       |
| Na₂O    | 9.31        |
| K₂O     | 6.40        |
| CaO     | 1.03        |
| Fe₂O₃   | 0.65        |
| MgO     | 0.12        |

3.1. Kinetic Experiments

The wt.% extraction of Al, Na and K as a function of time are shown in Figure 3a and the wt.% extraction of Si is presented in Figure 3b.
Figure 2. Powder X-ray diffractogram of nepheline syenite used in this study.

Figure 3. Effect of leaching retention time on the extraction of (a) Al, K, Na and (b) Si values (stirring rate: 300 rpm, HCl concentration: 5.89 M, temperature: 90 °C).

It is observed that time is not a crucial factor for Al, Na and K extraction under specific conditions. More specifically Al extraction is steady at near 50% from the first 30 min till 24 h retention time. Sodium is extracted at a substantially higher yield of 57% in relation to
K achieving only 15%. During the dissolution of aluminosilicate matrix, silicon is initially dissolved and reprecipitated as seen in Figure 3b.

3.2. Effect of Pulp Density on Extraction of Metals

The wt.% extraction of Al, Na and K, as a function of pulp density are provided in Figure 4a, and that of Si, in Figure 4b. The concentrations of Al, Na and K, expressed in g/L of produced pregnant liquid solutions as a function of pulp density, are provided in the secondary axis in Figure 4a, and that of Si, expressed in ppm, is presented in the secondary axis of Figure 4b. The data presented is for a retention time of 6 h and a leaching temperature of 90 °C.

![Recovery vs. Pulp Density](image)

**Figure 4.** Effect of pulp density on the extraction of (a) Al, K, Na and (b) Si values (pulp density: 2.5% w/v, stirring rate: 300 rpm, HCl concentration: 5.89 M, temperature: 90 °C).

In this leaching series, the stoichiometric excess of leaching agent led to similar extraction values for Al, Na and K. However, higher pulp density resulted in higher concentrations of Al, Na and K in solution (Figure 4a). Si concentration, as depicted in Figure 4b, increases slightly with increase of pulp density. Si high concentration in solutions justifies the silica gel formation, observed at higher pulp density values and longer retention time (more than 6 h) for pulp densities of 5 and 10% w/v.

Lower Si concentration, and avoidance of silica gel formation, was achieved with the application of HCl fuming tests. Table 2 presents the extracted values of Al, Na, K and Si from a conventional leaching test and a fuming test at a pulp density of 10%.

| Applied Method       | Al (g/L) | Na (g/L) | K (g/L) | Si (ppm) |
|----------------------|----------|----------|---------|----------|
| Conventional leaching| 5.66     | 4.02     | 0.79    | 77.6     |
| Fuming               | 5.57     | 4.00     | 0.64    | 21.18    |

As it is illustrated in Table 2, fuming technique led to a little lower dissolution of Si, and did not affect the concentrations and thus recoveries of the other metals. Another
benefit of the fuming method is the fast vacuum filtration separation of the PLS from the solid residues.

3.3. Effect of Temperature on Extraction of Metals

The wt.% extraction of Al, Na, K and Si as a function of temperature is shown in Figure 5a. From Figure 5a, it is noted that Al extraction is enhanced marginally by the raising of leaching temperature. Although, the impact of thermal intensification on Al extraction is low, as from 90 to 150 °C, the recovery increased from 51 wt.% to 62 wt.% only. A similar trend in the extraction curves is observed for K and Na. As shown in Figure 5b, Si extraction over the same temperature range varies from 1 to 0.60 wt.%. It is worth noting that Si extraction decreased with increased leaching temperature.

![Figure 5a](image1.png)  
![Figure 5b](image2.png)

**Figure 5.** Effect of temperature on the extraction of (a) Al, K, Na and (b) Si values (pulp density: 2.5% w/v, retention time: 6 h, stirring rate: 300 rpm, HCl concentration: 5.89 M).

3.4. Mineralogical Analysis of Leaching Residues

The XRD diffractograms of the residues at 90° and 150 °C in comparison with the raw material are presented in Figure 6. It is seen that the phases dissolved during leaching were exclusively nepheline and analcime which belong to the group of feldspathoids: natrolite which is a zeolite, and catapleite. These phases are dissolved substantially even at 90 °C and further temperature intensification simply hastens their dissolution. Na- and K-feldspars such as albite, microcline and anorthoclase are retained in the residue. Although the presence of amorphous silica is not directly evident in the XRD pattern, its presence is suggested by the decrease in the intensity of the peaks.
4. Discussion

Literature data collected and presented by Mase [19] states the reactivity of most common silicates with HCl solution. Nepheline, natrolite and analcime are mentioned and characterized as phases soluble to HCl aqueous solutions. Those data are in full agreement with the presented results, as nepheline, natrolite and analcime were easily and quickly dissolved, and finally disappeared from the leaching residues. The fast dissolution mechanism of nepheline, natrolite and analcime is also connected with silica gel formation [20]. As it has been reported, one of the possible results of the action of acids on a silicate is the complete breakdown of the structure followed by the dissolution of metal cations and silica [20]. Gelenatization of PLS would occur due to ready polymerization of silica in the aqueous solutions. Empirical rules, such as the aluminum to silicon mass ratio of framework silicates [21] and the framework charge ratio, expressed as Al: Al + Si mass ratio [22] are settled for the characterization of a silicate as prone to silica gel formation through an acid treatment. Based on such criteria, nepheline, natrolite and analcime are included in the list of gelatinizing minerals [21]. Nepheline and zeolites that were detected in our raw material have a high silicon content, in comparison with the other included mineralogical phases that would reasonably lead to silica gel formation and consequently create problems to the leaching process. Silica gelation phenomena were not observed in our leaching tests at the lowest studied pulp density of 2.5% w/v due to extremely high excess of HCl due to azeotropic solution. At higher pulp densities, gelation was observed in PLS at retention times higher than 6 h due to a substantially higher amount of silicon dissolved in the solution under the higher pulp densities.

Silica gel formation, observed at high pulp density, would be mitigated through the application of HCl fuming [23] approach. Indeed, the fuming test, presented in this work, led to similar extraction rates with conventional leaching without showing any indication of silica gelation.

The low % extraction yield of Al, estimated to be near 50%, calculated in all leaching tests performed at a temperature of 90 °C, could be translated in that 50% of the contained Al was not dissolved. Based on the results of residue characterization, the undissolved Al is in the form of feldspars, which is in agreement with the literature [19]. The increase on leaching temperature did not improve appreciably the Al extraction yield. Under acidic conditions, alkali feldspars’ dissolution in aqueous solutions is incongruent. Reports on
the incongruent dissolution of alkali feldspars and their dissolution kinetics at a variety of examined temperatures [24,25] concluded that the increase in the molar concentration of alkali ions might be faster than that in the molar concentration of aluminum. This observation could be attributed to the preferable leachability of alkali ions from the feldspar matrix, or to the precipitation of an aluminum-containing phase. Different models of incongruent feldspar dissolution have been proposed and led to different routes of rate control in dissolution. Among them, two approaches proposed that dissolution rate is controlled by the inability of feldspar components to diffuse through a protective layer that is formed around feldspar grains [26]. This layer could have the form of an amorphous silica-alumina precipitate [27] or a crystalline precipitate whose phases are based on the chemical content of the solution [28].

Incongruent dissolution in the matrix of nepheline syenite is also endorsed by the comparison of dissolution curves of Al, K and Na, presented in Figure 5a. K and Na dissolution seems to follow the trend of Al dissolution as a slow increase to the extraction yields with the increase of temperature is noted. Potassium, in addition, achieved the lowest extraction yield of the order of 15–20%, as the list of soluble minerals that were dissolved during the leaching process do not have potassium content. Potassium exists only in feldspars and especially in anorthoclase which is the second most abundant mineral in the raw material and in a subordinate amount in microcline. The presence of low amounts of K in solution thus suggests low level dissolution of feldspars in agreement with the literature. Moreover, the relative high amount of dissolved sodium reached a yield of 73% at 150 °C relative to aluminium of 62% at 150 °C which (a) strengthened the above observation for initiation of the feldspars dissolution process and (b) seemed to support the incongruent dissolution mechanism of feldspars.

Finally, although the leaching process with azeotropic HCl seems to be inefficient due to (a) the difficulties which exist in dissolving the alkali feldspars and (b) the silica gelation phenomena observed at high pulp densities, the application of HCl fuming approach, that can solve the latter but not the former drawback, has the potential to produce pregnant solutions with high aluminum content proper for alumina production. Furthermore, the produced residue that contains exclusively undissolved feldspars and amorphous silica is an attractive, and already used raw material for glass, ceramic and other industries [29].

5. Conclusions

In this work, the use of a mine waste product, characterized as nepheline syenite, as an alternative alumina source in a leaching process with azeotropic HCl solution, is examined. Leaching of nepheline syenite with azeotropic HCl acid yielded a PLS with 55% wt. Al extraction at 90 °C. Increasing the leaching temperature to 150 °C resulted in only a marginal increase in the amount of dissolved Al. This low Al extraction could be attributed to the low dissolution rate of alkali feldspars, under the experimental conditions. However, techniques presented in this work, and more particularly HCl fuming, promise a feasible route for the production of a more concentrated Al solution with the absence of silica gelation. Moreover, by this process a residue, containing feldspars which are not dissolved in the fuming process, is produced that could be feed for glass and ceramic industries. Therefore, from this point of view, this route after optimization could offer a potentially viable approach for holistic utilization of such a type of mine waste.

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References
1. Frank, W.B.; Warren, E.H.; Robert, K.D.; Douglas, A.G.; Maurice, W.W.; Kenneth, J.C.; Thomas, B.B. Aluminum. In Ullmann’s Encyclopedia of Industrial Chemistry: Wiley-VCH: Weinheim, Germany, 2007.
2. Sverdrup, H.U.; Ragnarsdottir, K.V.; Kocac, D. Aluminium for the future: Modelling the global production, market supply, demand, price and long term development of the global reserves. Resour. Conserv. Recycl. 2015, 103, 139–154. [CrossRef]
3. Allwood, J.M.; Cullen, J.M.; Carruth, M.A.; Cooper, D.R.; McBrien, M.; Milford, R.L.; Moynihan, M.C.; Patel, A.C. Sustainable Materials with Both Eyes Open; UIT Cambridge Ltd.: Cambridge, UK, 2012.
4. Habashi, F. Handbook of Extracitive Metallurgy; Wiley-VCH: Weinheim, Germany, 1998; Volume 2.
5. Meyer, F.M. Availability of Bauxite Reserves. Nat. Resour. Res. 2004, 13, 161–172. [CrossRef]
6. Blengini, G.; Latunussa, C.E.L.; Eynard, U.; de Matos, C.T.; Wittmer, D.; Georgitzikis, K.; Pavel, C.; Carrara, S.; Mancini, L.; Unguru, M.; et al. Study on the EU’s List of Critical Raw Materials (2020) Final Report; European Commission: Brussels, Belgium, 2020.
7. Gaudernack, B.; Gjelsvik, N.; Farbu, L. Process for the Extraction of Alumina from Aluminum-Containing Silicates. U.S. Patent 4,110,399, 29 August 1978.
8. Boudreault, R.; Alex, S.; Biasotto, F. Processes for Extracting Aluminum from Aluminous Ores. U.S. Patent 8337789, 7 May 2008.
9. Aranda, A.; Mastin, J. Alumina and Carbonate Production Method from Al-Rich Materials with Integrated CO2 Utilization. U.S. Patent 9,963,352, 8 May 2015.
10. Coenen, A.; Kosswig, K.; Hentschel, B.; Ziebarth, J.; Huels, A.G. Method of Manufacturing Hydrogen Chloride from Solutions of Amine Hydrochlorides. U.S. Patent 4,230,681, 28 October 1980.
11. Baniel, A.; Hansen, R.P. Process for the Recovery of Hydrochloric Acid. U.S. Patent 2012/0134912, 31 May 2012.
12. Baniel, A.; Eyal, A. Process for the Recovery of HCL from a Dilute Solution Thereof and Extractant Composition for Use Therein. U.S. Patent 2011/0028710, 3 February 2011.
13. Jena, S.K.; Dhawan, N.; Rao, D.S.; Misra, P.K.; Mishra, B.K.; Das, B. Studies on extraction of potassium values from nepheline syenite. Int. J. Miner. Process. 2014, 133, 13–22. [CrossRef]
14. Abouzeid, A.-Z.M.; Negm, A.-T.A. Characterization and beneficiation of an Egyptian nepheline syenite ore. Int. J. Mineral. 2014, 128246. [CrossRef]
15. Saller, M. In a state of flux- feldspar and nepheline syenite reviewed. Ind. Miner. 1999, 385, 43.
16. Jena, S.K.; Dhawan, N.; Rath, S.S.; Rao, D.S.; Das, B. Investigation of microwave roasting for potash extraction from nepheline syenite. Sep. Purif. Technol. 2016, 161, 104–111. [CrossRef]
17. Terry, B. The acid decomposition of silicate minerals part I. Reactivities and modes of dissolution of silicates. Hydrometallurgy 1983, 10, 135–150. [CrossRef]
18. Danjun, T.A.; Hongwen, M.A.; Ge, L.I.; Hao, L.I.; Dan, Z.O. Sintering Reaction of Pseudoleucite Syenite: Thermodynamic Analysis and Process Evaluation. Earth Sci. Front. 2009, 16, 269–276.
19. Hanzo, M. On the Reactivity of Various Silicate Minerals toward Acids. Bull. Chem. Soc. Jpn. 1961, 34, 214–225.
20. Terry, B. The acid decomposition of silicate minerals part II. Hydrometallurgical applications. Hydrometallurgy 1983, 10, 151–171. [CrossRef]
21. Murata, K.J. Internal structure of silicate minerals that gelatinize with acid. Am. Mineral. 1943, 28, 545–562.
22. Marshall, C.E. The Physical Chemistry and Mineralogy of Soils. Volume I: Soil Materials. Soil Sci. 1965, 99, 355. [CrossRef]
23. Davis, P.; Stopic, S.; Balomenos, E.; Panias, D.; Paspalarias, I.; Friedrich, B. Leaching of rare earth elements from eudialyte concentrate by suppressing silica gel formation. Miner. Eng. 2017, 108, 115–122. [CrossRef]
24. Wollast, R. Kinetics of the alteration of K-feldspar in buffered solutions at low temperature. Geochim. Cosmochim. Acta 1967, 31, 635–648. [CrossRef]
25. Lagache, M. Contribution à l’étude de l’altération des feldspaths, dans l’eau entre 100 et 200 °C, sous diverses pressions de CO2 et application à la synthèse des minéraux argileux. Bull. Minéral. 1965, 88, 223–253. [CrossRef]
26. Petrović, R.; Berner, R.A.; Goldhaber, M.B. Rate control in dissolution of alkali feldspars—I. Study of residual feldspar grains by X-ray photoelectron spectroscopy. Geochim. Cosmochim. Acta 1976, 40, 537–548. [CrossRef]
27. Correns, C.W.; von Engelhardt, W. Neue Untersuchungen über die Verwitterung des Kalifeldspates. Naturwissenschaften 1938, 26, 137–138. [CrossRef]
28. Helgeson, H.C. Kinetics of mass transfer among silicates and aqueous solutions. Geochim. Cosmochim. Acta 1971, 35, 421–469. [CrossRef]
29. Apodaca, L.E. Feldspar and nepheline syenite. J. U. S. Geol. Surv. Miner. Yearb. 2008, 1, 3057.