

Article

High-Iron Bauxite Residue (Red Mud) Valorisation Using Hydrochemical Conversion of Goethite to Magnetite

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Abstract: Bauxite residue (BR), also known as red mud, is a by-product of the production of alumina via the Bayer process. Because of the high sodium oxide and other impurities content, this material is not used to obtain iron or other iron-containing products. In this paper, the hydro-chemical conversion of goethite (FeOOH) to magnetite (Fe₃O₄) in high-iron BR from the Friguia alumina refinery (Guinea) by Fe²⁺ ions in highly concentrated alkaline media was studied. The simultaneous extraction of Al and Na made it possible to obtain a product containing more than 96% Fe₃O₄. The results show that the magnetization of Al-goethite and Al-hemetite accelerates the dissolution of the Al from the iron mineral solid matrix and from the desilication product (DSP). After ferrous sulfate (FeSO₄·7H₂O) was added directly at the FeO:Fe₂O₃ molar ratio of 1:1 at 120 °C for 150 min in the solution with the 360 g L⁻¹ Na₂O concentration, the alumina extraction ratio reached 96.27% for the coarse bauxite residue size fraction (Sands) and 87.06% for fine BR obtained from red mud. The grade of iron (total iron in the form of iron element) in the residue can be increased to 69.55% for Sands and 58.31% for BR. The solid residues obtained after leaching were studied by XRD, XRF, TG-DTA, VSM, Mössbauer spectroscopy and SEM to evaluate the conversion and leaching mechanisms and the recovery ratio of Al from different minerals. The iron-rich residues can be used in the steel industry or as a pigment.

Keywords: alkali leaching; red mud; high-iron waste; disposal; Al-goethite; conversion; magnetite

1. Introduction

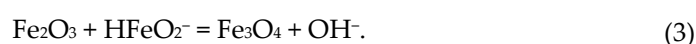
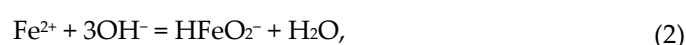
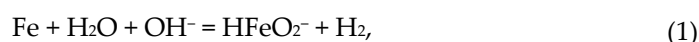
BR is a solid fraction of the red mud. Each year, over 150 million tons of red mud are produced during the Bayer process, which is used to extract Al from bauxites by the alkaline method [1]. Because of this method, red mud has a high amount of caustic alkali and other impurities, which makes it very toxic [2]. Moreover, the concentration of heavy metals in BR can be doubled after alumina extraction [3,4]. The utilization rate of BR is very low because of its high toxicity and corrosivity [5–7].

BR has a high number of valuable components, such as iron, rare earth elements (REEs), vanadium, titanium, etc. [8–14]. The extraction of these elements from BR together with the reduction of its toxicity can be economically and ecologically beneficial. For example, BR produced from the Fria bauxite deposit in Guinea has a very high amount of iron: 40% of the iron in red mud obtained after digestion of bauxites on the Friguia

alumina refinery, or 55-60% of the iron in the sands obtained by the gravity method of separation of BR before thickening and washing step [15].

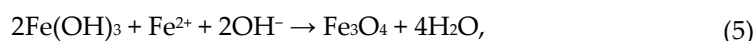
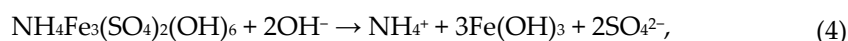
Different methods of iron production from BR have been studied so far. These methods can be divided into pyro- and hydrometallurgical [16]. Pyrometallurgical methods include magnetic separation after preliminary reductive roasting [17], reductive sintering with different fluxes [18,19], and smelting BR with a reducing agent to make pig iron [20,21]. The major disadvantage of these methods is their high energy consumption, since the temperatures of these processes can reach as high as 1000-1750 °C [22].

Li et al. [23,24] and Pasechnik et al. [25] showed that hematite from BR can be converted into magnetite during a hydrothermal reduction in the presence of iron powder, Fe²⁺, and OH⁻ ions (Equation (1)-(3)). However, a high-pressure process with a leaching temperature greater than 250 °C is required for complete iron conversion into magnetite.



Furthermore, Zhou et al. [26] showed that the addition of Al during high-temperature digestion of high-iron bauxite not only transforms iron into hematite covered by magnetite but also accelerates Al extraction from Al-goethite that is hard to dissolve by the Bayer method. However, a high-pressure leaching at 270 °C is required for the formation of the iron-enriched residue with an iron content of 56%. Such iron content is still very low, especially if the pigment-quality magnetite is proposed as a final product.

Vu et al. [27] and Bohacek et al. [28] proposed a method for the production of black pigment-quality magnetite from jarosite (NH₄Fe₃(SO₄)₂(OH)₆) by decomposition in an ammonia or sodium alkali environment in the presence of ferrous sulfate (Equations (4) and (5)). The mixture was then heated to 90 °C and maintained for 7 h to obtain a well-crystallized product with iron content higher than 70% [27].



This freshly precipitated Fe(OH)₃ is very reactive, so magnetite can be formed at a lower temperature (90 °C vs. 250 °C for hematite in bauxites). Hage et al. [29] suggested a method for making magnetite from jarosite using MgO as a neutralizing agent and acid-treated cellulose as a reductant. Using a reducing agent saves expensive alkali, which is neutralized and turned into Na₂SO₄ by adding FeSO₄. Both methods allow for the obtainment of high-quality magnetite products from jarosite that itself has very low impurities.

In this work, a novel method of atmospheric pressure leaching (APL) of Guineas high-iron BR in the presence of FeSO₄ was proposed, which led to almost complete dissolution of Al from the solid matrix of the sands with the formation of the pigment-quality magnetite containing about 70% of Fe. The extraction of Al from the red mud was also higher than 87%, however, the iron content in the solid residue reached only 58 %. The optimum conditions of alumina extraction and iron minerals conversion as well as the mechanism of these processes were verified. The solids were examined using X-ray diffraction (XRD), X-ray fluorescence (XRF), Scanning electron microscopy with energy dispersive spectroscopy (SEM-EDS), Mossbauer spectrometry, Vibrating sample magnetization (VSM), and spectrophotometry methods.

2. Materials and Methods

2.1. Chemical composition of samples

The chemical composition of the raw BR and residues after leaching were analyzed using a powder X-ray fluorescence spectrometry with an Axios MAX spectrometer (Malvern Panalytical Ltd., Almelo, Netherlands). Tablet-shaped samples (\varnothing 20 mm; 300 mg) were prepared for analysis via pressing. Also, polystyrene was used as a binder at a ratio of 5:1. Super Q software was used to calculate the metallic contents of the samples.

The Sc content in BR and sands leaching residues were determined by inductively coupled plasma mass spectrometry (ICP-MS) using an X Series II quadrupole mass spectrometer (Thermo Scientific, Germany). The quantitative determination of the Sc was carried out using iPlasmaProQuad software (GEOKHI RAS, Moscow, Russia)

2.2. Physical properties of samples

The phase composition of the raw BR and residues after leaching was analyzed using X-ray diffraction using a Difrei-401 diffractometer (JSC Scientific Instruments, Saint Petersburg, Russia) using a Cr-K α radiator with 2θ angles ranging from 15° to 140°. The X-ray source operated with an output of 25 kW and an exposure time of 30 min. The diffraction data were processed using the Match 3! (Crystal Impact GmbH, Bohn, Germany) software. The particle size distribution of the samples was measured with a SALD-2201 Laser Diffraction Particle Size Analyser (Shimadzu, Kyoto, Japan). The surface morphology and elemental mapping of the raw materials and the residues were investigated using SEM-EDX methods on a Vega III microscope (Tescan, Brno, Czech Republic). The Mössbauer analysis of raw BR and sands, residues after leaching were done using a spectrometer MS1104EM (Cordon, Rostov-on-Don, Russia) at 296 \pm 3 K and 77.6 \pm 0.5 K temperature in a vacuum cryostat. The ^{57}Co nuclei in Rh matrix with 3 mCi activity (RITVERC, Saint Petersburg, Russia) was used as the γ -radiation source. The SpectrRelax 2.8 software (Lomonosov Moscow State University, Moscow, Russia) was used to analyze the Mössbauer spectra. Values of chemical shifts are presented relative to α -Fe. The magnetic measurements were performed using a MPMS-XL-7 SQUID magnetometer (Quantum Design, San Diego, California, USA). The TG-DTA analysis was performed using simultaneous thermogravimetry analyzer STA 449 F3 Jupiter (Netzsch, Selb, Germany). over a range of 33–1100 °C at a heating rate of 10 °C min $^{-1}$. An air atmosphere was used, with a flow rate of 20 ml min $^{-1}$.

2.3. Materials

The bauxite residue and sands used in this research were obtained from Friguia alumina refinery in Guinea with GPS coordinates N10.39°, W13.58°. Sands are the coarser particles of residue obtained from red mud by separation through hydrocyclones before the thickening and washing step. The chemical compositions of BR and the sands are presented in Table 1. Figure 1 shows the particle size distribution of samples. It is obvious that sands obtained using a hydrocyclone separator are significantly coarser than BR, and the particle size distribution of the sands is more uniform. Figure 2 shows the morphology and elemental mapping of the samples. BR is composed of particles smaller than 10 μm , and the Fe, Na, Si, and Al elemental mapping are similar, which can be explained by the precipitation of DSP and gibbsite during the thickening and washing steps. The sands elemental map shows that there are a lot of goethite particles and some single aluminum particles with a small amount of silicon and sodium and a lot of iron on the surface. This implies that desilication of the solution is ending during the thickening and washing stage. The re-precipitation of gibbsite from a diluted aluminate solution during the washing step also leads to a higher concentration of aluminum in the BR.

The other reagents, NaOH and FeSO $_4$ ·7H $_2$ O, were of analytical grade. Sodium alkaline solutions with a Na $_2$ O concentration of 330-400 g L $^{-1}$ were obtained by mixing NaOH with distilled water.

Table 1. The chemical compositions of bauxite residue (BR) and sands from Friguia alumina refinery, Guinea, wt. %.

Sample	Fe	Si	Ti	Al	Na	O	Other
BR	41.7	4.25	4.36	12.34	4.53	30.68	2.14
Sands	56.23	0.77	1.38	5.55	1.26	33.80	1.01

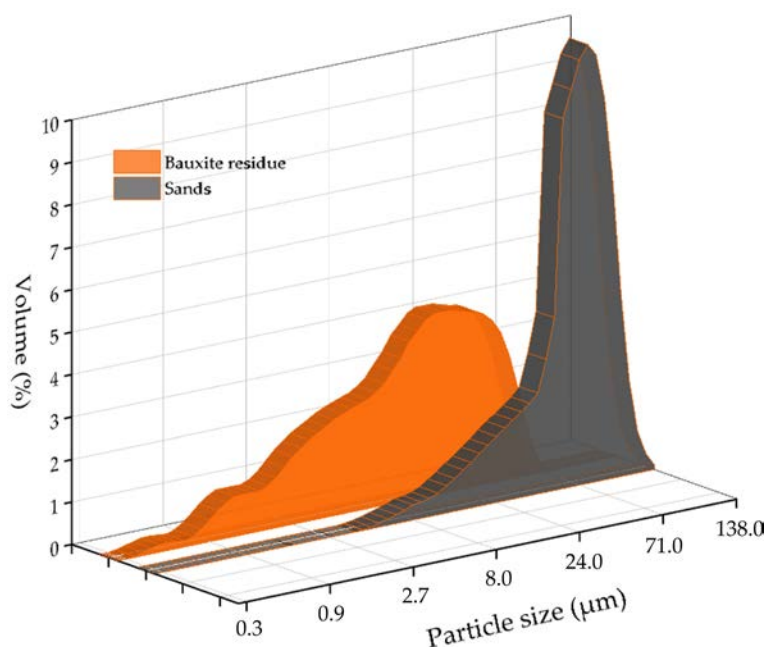


Figure 1. The particle size distribution of the bauxite residue (BR) and sands from the Friguia alumina refinery.

2.4. Experimental

The extraction of Al with NaOH and the conversion of goethite to magnetite were carried out in the thermostated 0.5 L stainless steel reactor. The reactor has openings for stirring as well as for controlling temperature and recycling evaporated water through a water-cooled condenser. The L:S ratio and the stirring speed in all experiments were 100 g L⁻¹ and 300 rpm, respectively. The molar ratio of Fe²⁺ added in the form of iron sulphate to Fe₂O₃ in BR or sands according to Equation (3) was equal to 1.0. The BR, sands, and iron sulphate were added to a hot solution with a Na₂O concentration of 330 g L⁻¹ to 400 g L⁻¹. This made it possible to use the atmospheric leaching process at a temperature of 120 °C. The temperature of the leaching was varied from 100 to 120 °C, and the leaching time from 1 to 5 h. After leaching, the pulp was filtered, and the solid residue was dried at 110 °C for 8 h before analysis.

2.5. Experimental data calculation

In order to avoid the mutual influence of factors on each other and to reduce the number of experiments, a Box-Benken experimental design created in the Statistica 13 software (TIBCO, Hamburg, Germany) was used in this research. The design consists of three blocks of fifteen experiments each, with varying parameters at three levels. The output parameters are the extraction of aluminum into solution and the concentration of iron in the solid residue.

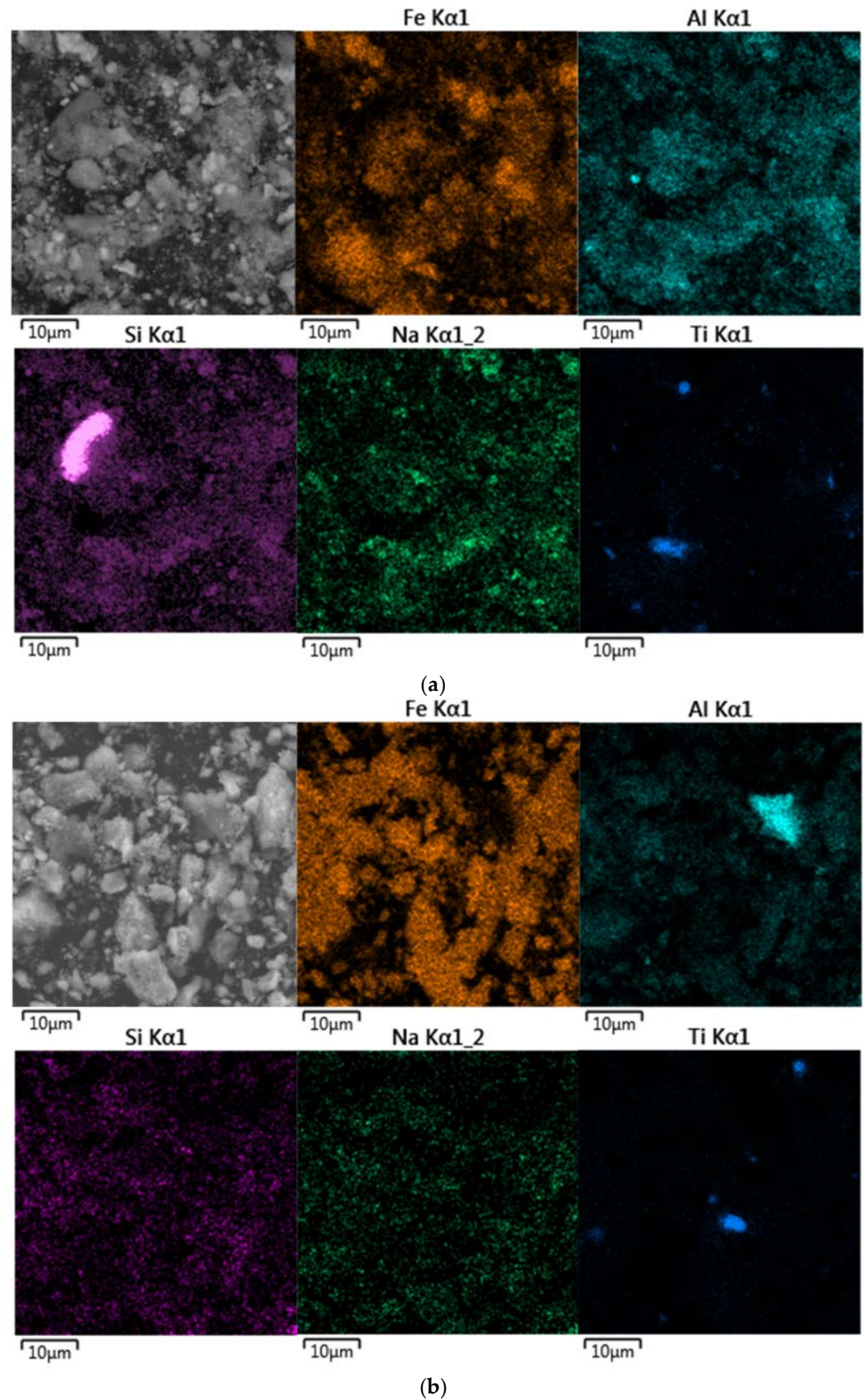


Figure 2. BSE image and elemental mapping of the BR particles surface (a); BSE images and elemental mapping of the sands particles surface (b).

The statistical-based automated neural network (SANN) was used for modeling the Al and Fe extraction and Na_2O concentration in the solid residue. SANN is an artificial

intelligence-based method that adjusts the result of modeling until the desired quality is obtained. Statistica 13 was used to make SANN models using a multilayer perceptron (MLP) method.

The amount of Al that was extracted from the different aluminum minerals (X) into the solution was estimated using Equation (6):

$$X = (m_1 \times X_1 - m_2 \times X_2) / (m_1 \times X_1), \quad (6)$$

where m_1 – the mass of the original sample, [g]; X_1 – mineral content in the original sample, [%], m_2 – the mass of the leaching residue, [g]; X_2 – mineral content in the leaching residue, [%].

To assess the content of Na in the raw materials and solid residue, we used the assumption that all the Na in the product is represented by DSP (equation (7):

$$DSP = [(Na \times 23) / (27 \times Al)] \times 100\%, \quad (7)$$

where Na – the content of sodium in a sample, [%]; Al – the Al content in a sample, [%]; 23/27 – the molar ratio of Na vs Al in the sodalite.

In the case of sands that have only two Al minerals, the equation (8) was used to calculate the aluminum contained in solid matrix:

$$\text{Solid matrix}_{\text{sands}} = [(Al - DSP) / Al] \times 100\%, \quad (8)$$

where Al – the Al content [20] in a sample, [%]; DSP – the amount of Al in the DSP in this sample calculated by equation (7), [%].

To determine the amount of aluminum hydroxides in the solid matrix of BR iron minerals, we assumed that the amount of this Al has a linear dependence on the Fe content in the residue (Equation (9):

$$\text{Solid matrix}_{\text{BR}} = [(\text{Solid matrix}_{\text{sands}} \times \text{Fe}_{\text{BR}}) / \text{Fe}_{\text{sands}}] \times 100\%, \quad (9)$$

where $\text{Solid matrix}_{\text{sands}}$ – the amount of Al in the solid matrix of the raw sands iron minerals calculated by equation (8), [%]; Fe_{BR} – the Fe content in the BR, Fe_{sands} – the Fe content in the raw sands, [%].

Therefore, all the remaining Al in the BR is represented by the reprecipitated gibbsite.

The blackness (jetness) evaluation was made using an SR-60 colorimeter (Shenzhen ThreeNH Technology Co., Shenzhen, China). To obtain a single numerical result, the calculation of the magnitude of the blackness (M_y) was based on one of the values of the color coordinates - L (brightness) that was determined by Equation (10):

$$M_y = 100 \times \log(100 / L). \quad (10)$$

This only determines how light or dark the sample is without taking the color shade into account. Since the shade of black affects how dark something looks, the degree of jetness (M_c) was determined by changing CIE LAB results to CMYK in the color model with the following equation (11):

$$M_c = 100 \times ((\log(X_n / X) - \log(Y_n / Y) + \log(Z_n / Z))). \quad (11)$$

The M_c value describes a higher blackness if the shade is blue (Z), and a lower saturation if the shade is yellow (Y). X_n , Y_n , Z_n - standard color values of the illuminating light type.

3. Results and discussion

3.1. The nature of the Al-containing phase in the raw materials

Previous research by Zhou et al. [26] showed that the rate of Al extraction from Guinea's bauxite with a high iron content is limited by the presence of aluminum substituted goethite (Al-goethite), which is refractory using conventional Bayer process conditions. To study the occurrence of Al in the BR and the Sands from the Friguia alumina refinery, the phase composition of the samples was studied using XRD, TGA and Mössbauer spectrometry. The XRD patterns of the BR and the Sands are shown in Figure 3. Figures 6a and 6b indicate that there are differences in hematite, goethite and gibbsite content in the raw BR and sands. The peaks of goethite are higher in the sand samples, whereas hematite is found in higher concentrations in the BR. Additionally, there are peaks of sodalite (DSP) in the BR that are not seen in the sands due to the low content.



Figure 3. The XRD patterns of the BR and the solid residue obtained after BR leaching at the optimal conditions (a); the XRD patterns of the Sands and the solid residue obtained after Sands leaching at the optimal conditions (b).

Figure 4 shows the TG-DTA curves of the BR and the Sands. There are significant differences in the DTA of these two samples. Figure 4a shows that the decomposition of BR occurred in five stages, similar to the results obtained by Zhou et al. [30] for Al-goethite containing bauxite residues. The 1st stage (from 25 to 220 °C) is attributed to the loss of moisture [21]. The 2nd stage (from 220 to 330 °C) with the endothermic peak at 299.1 °C is associated with the decomposition of gibbsite [31]. The 3rd stage occurred at temperatures between 330 and 370 °C, and an endothermic peak of goethite and al-goethite decomposition was found at 335.5 °C [32]. Between 335 and 420 °C the 4th stage, the boehmite undergoes decomposition. And the 5th stage with an endothermic peak of hydrosodalite decomposition was found at 495 °C [33]. The TG-DTA curve of the sands shows only endothermic peaks of goethite at 344.3 °C and a much smaller peak of hydrosodalite at 504.8 °C. These findings indicate that there is practically no gibbsite in the Sands and all aluminum is enclosed in a solid matrix of minerals and DSP.

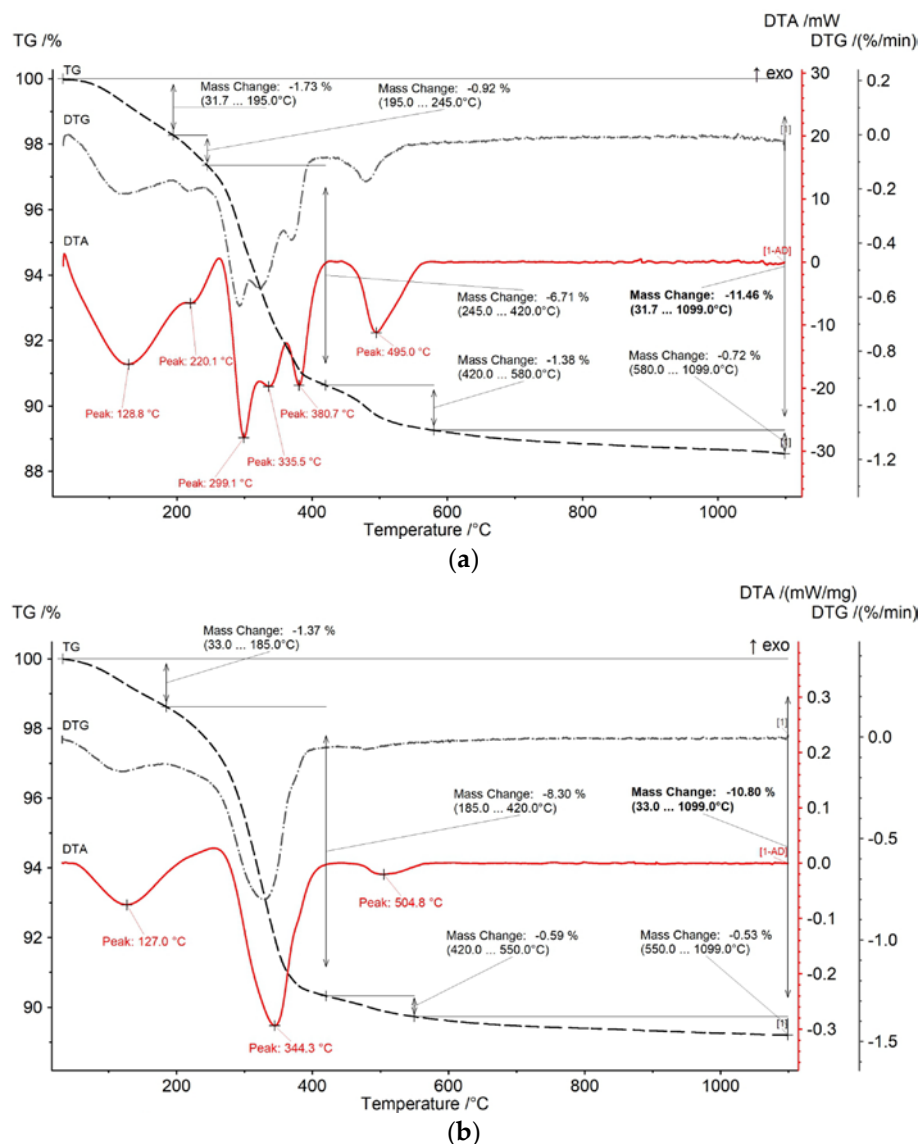


Figure 4. The TG-DTA curves of the BR (a); and Sands (b) at different temperatures.

Figure 5 shows that the results of the Mössbauer analysis of raw BR and Sands. - Samples at room temperature have a similar profile and are combinations of an intense paramagnetic doublet and a sextet with fairly narrow resonance lines (Figure 5). A noticeable difference between the spectrum of the Sands sample and BR is that the first sample has extended absorption between the first and fifth lines of the sextet (Figure 5a and 5c). At the boiling point of nitrogen, the shape of the spectra changes significantly: the doublet almost completely disappears, and the outer lines of the narrow sextet split in pairs (Figure 5b and 5d). All this testifies to the complex composition of the materials.

The Mössbauer spectra of the raw materials can be described by a superposition of four sextets and two doublets (Figure 5, Table 2). Two outer sextets at 296 K and one outer sextet at 78 K correspond to aluminum-substituted hematite - α -(Fe_{1-x}Al_x)₂O₃ [34]. The low value of magnetic splitting even for frozen samples, and the absence of the Morin transition (TM=260 K for α -Fe₂O₃) do not allow us to attribute these subspectra to pure unsubstituted hematite - α -Fe₂O₃ [22,35]. The isomorphic substitution of aluminum for iron atoms [36] leads to a decrease in the effective magnetic field recorded in the spectra by approximately 1 kOe by 1 w.% for 300 K [37,38]. Analyzing the data presented in [38], it can be shown that the degree of substitution x in alumina hematite - α -(Fe_{1-x}Al_x)₂O₃ can be expressed in terms of the effective magnetic field at room temperature, as:

$$x = 5.4982 - 0.0107 * H_{eff}, \quad (12)$$

Then, based on the experimental data (Table 2) for the studied samples using Equation (12), it is possible to determine the composition of Al-hematite as $\alpha\text{-(Fe}_{0.9}\text{Al}_{0.1})_2\text{O}_3$ for the 1st subspectrum, and $\alpha\text{-(Fe}_{0.7}\text{Al}_{0.3})_2\text{O}_3$ for the 2nd subspectrum (Table 2).

The next pair of sextets poorly resolved at room temperature, upon cooling to the boiling point of nitrogen, transforms into a pair of well-defined sextets with parameters close to goethite [39]. The reason for the low value of the effective magnetic field even at low temperatures is the partial replacement of iron atoms by aluminum, with the formation of Al-goethite – $\alpha\text{-Fe}_{1-x}\text{Al}_x\text{OOH}$, for which the temperature transformations of the spectrum are discussed in detail in [16].

The paramagnetic part of the high-temperature Mössbauer spectra may have a dual nature. On the one hand, 7-10% of the area of this pair of doublets can be related to the superparamagnetic fractions of the above described Al-hematite and Al-goethite [40]. The remaining part, as shown in [16] refers to akageneite substituted by aluminum – $\beta\text{-Fe}_{1-x}\text{Al}_x\text{O(OH, Cl)}$, which, when cooled to the boiling point of nitrogen transforms into a broadened sextet (Table 2). The remaining minor components in the low-temperature spectra in the form of doublets correspond to Fe^{3+} ions in the high-spin state and octahedral oxygen environment [41], which can isomorphically replace Al^{3+} ions [36] in the crystal lattice of, for example, sodalite [42].

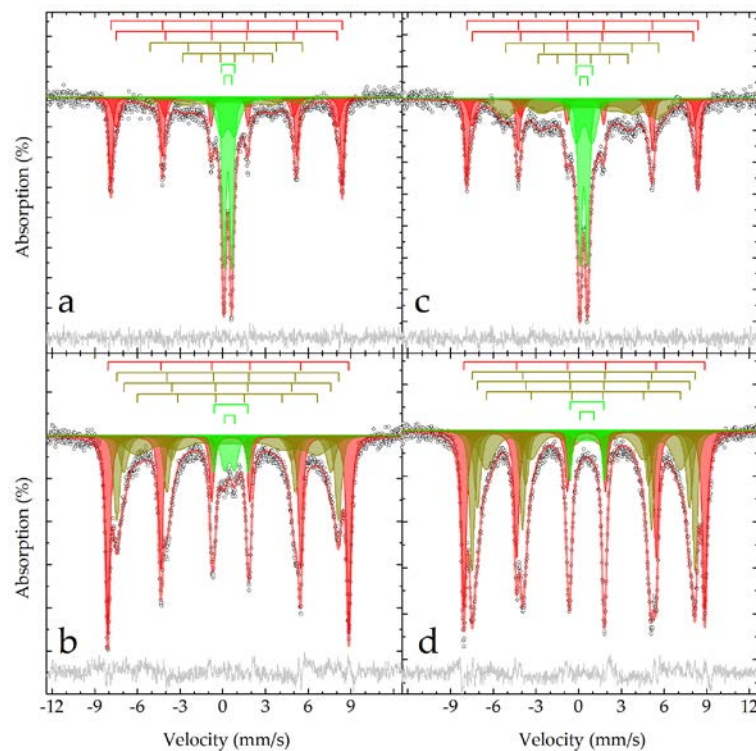


Figure 5. Mössbauer spectra obtained at 296 K (a, c) and 77.6 K (b, d) of the BR (a, b) and Sands (c, d) samples and their model description according to Table 2.

Based on the chemical analysis and the phase composition of the raw materials, the distribution of aluminum by minerals was calculated. Table 3 shows the results of the semi-quantitative calculation of the aluminum containing phase composition.

Table 2. The subspectrum parameters describing experimental Mössbauer spectra obtained at different temperatures for BR and Sands samples.

Sample		BR					Sands					
Temperature, K	Phase	№	δ	$\varepsilon (\Delta=2\varepsilon)$	Γ_{exp}	H_{eff}	S	δ	$\varepsilon (\Delta=2\varepsilon)$	Γ_{exp}	H_{eff}	S
			mm/s		kOe		%	mm/s		kOe		%
296	$\alpha\text{-(Fe}_{1-x}\text{Al}_x)_2\text{O}_3$	1	0.37±0.01	-0.10±0.01	0.34±0.01	504.7±0.2	26±1	0.37±0.01	-0.10±0.01	0.36±0.01	504.1±0.3	19±1
		2	0.37±0.01	-0.09±0.01	0.59±0.03	482±1	15±1	0.36±0.01	-0.08±0.01	0.58±0.04	482±2	11±1
	$\alpha\text{-Fe}_{1-x}\text{Al}_x\text{OOH}$	3	0.45±0.01	-0.21±0.01	1.60±0.01	332.1±0.1	10±1	0.45±0.05	-0.21±0.05	1.60±0.09	332±2	23±1
		4	0.32±0.01	-0.01±0.01	1.60±0.01	195.5±0.1	11±2	0.32±0.07	-0.01±0.05	1.60±0.09	196±4	15±1
	$\beta\text{-Fe}_{1-x}\text{Al}_x\text{O(OH, Cl)}$	5	0.37±0.01	(1.0±0.3)	0.88±0.07		15±5	0.39±0.01	(1.1±0.1)	0.84±0.09		12±2
		6	0.36±0.01	(0.53±0.01)	0.33±0.03		23±5	0.36±0.01	(0.52±0.01)	0.36±0.01		19±2
77.6	$\alpha\text{-(Fe}_{1-x}\text{Al}_x)_2\text{O}_3$	1	0.48±0.01	-0.08±0.01	0.37±0.01	525.2±0.1	38.4±0.7	0.47±0.01	-0.07±0.01	0.40±0.01	525.1±0.1	31.8±0.5
		2	0.47±0.01	-0.12±0.01	0.56±0.01	482.9±0.4	22±1	0.47±0.01	-0.12±0.01	0.43±0.01	485.5±0.2	24.9±0.9
	$\alpha\text{-Fe}_{1-x}\text{Al}_x\text{OOH}$	3	0.47±0.01	-0.15±0.01	0.75±0.04	451±1	13±1	0.47±0.01	-0.13±0.01	0.56±0.02	461.8±0.5	18±1
		4	0.40±0.04	-0.09±0.03	2.18±0.01	391.6±0.1	19.0±0.5	0.47±0.01	-0.14±0.01	1.34±0.04	422±2	21±1
	$\text{Fe}^{3+}_{\text{OH}} \text{ (DSP)}$	5	0.59±0.01	(2.37±0.01)	0.33±0.02		2.9±0.1	0.59±0.01	(2.38±0.01)	0.25±0.01		2.5±0.2
		6	0.47±0.01	(0.71±0.02)	0.62±0.03		4.8±0.2	0.57±0.05	(1.0±0.1)	1.2±0.3		2.0±0.3

δ – Isomer shift; $\varepsilon (\Delta)$ – Quadrupole shift (splitting); Γ_{exp} – Line width; H_{eff} – Hyperfine magnetic field; S – Relative area of a subspectrum №.

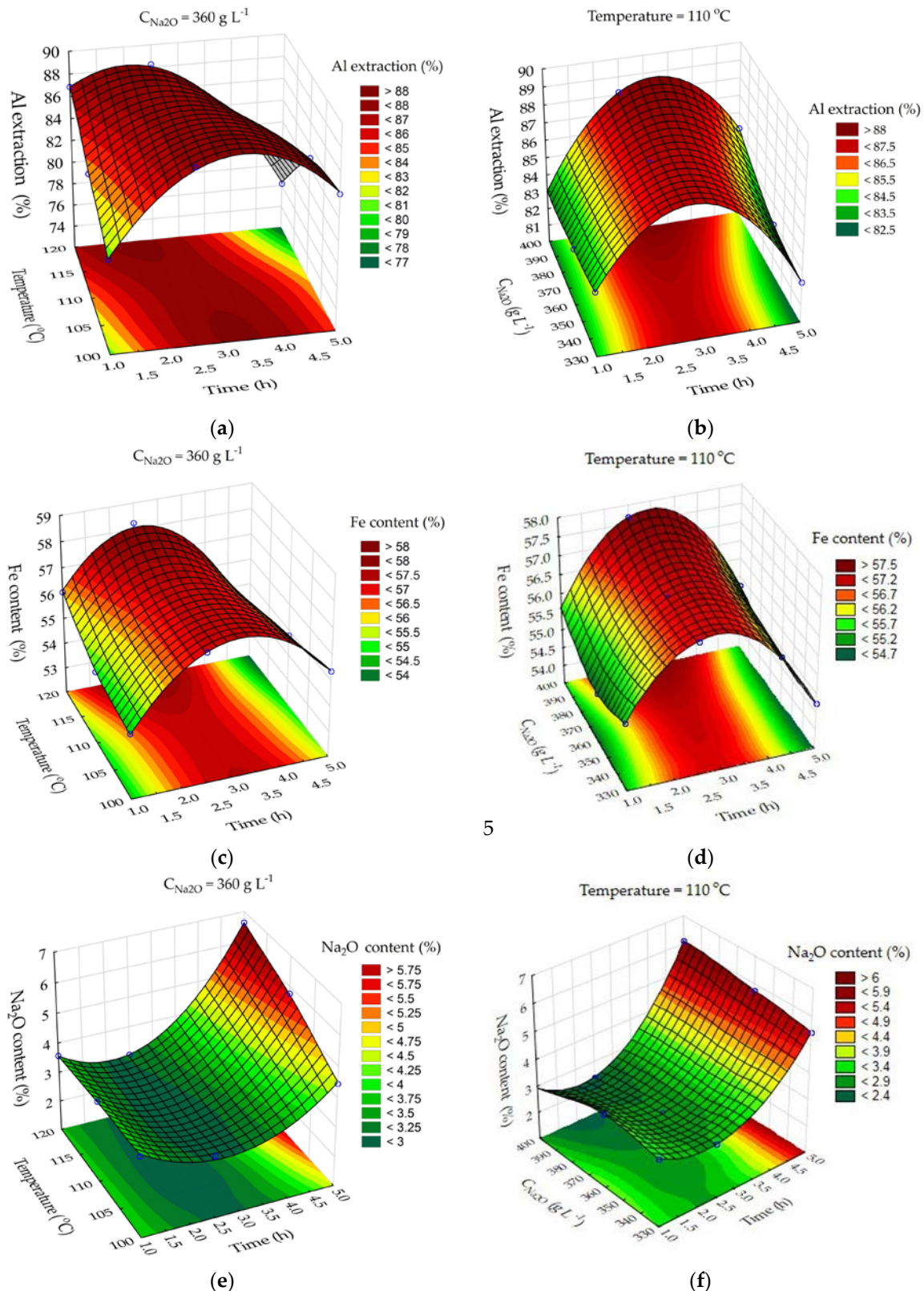
Table 3. The results of the semi-quantitative calculation of the aluminum containing phase composition in the BR and the sands, wt. %.

Sample	Desilication product	Aluminum hydroxide in the solid matrix	Re-precipitated gibbsite
BR	31.3	26.7	42.0
Sands	19.3	80.7	-

3.2. The effect of leaching parameters on the Al extraction and the Fe, Na content in the solid residue

To see how different leaching parameters affect the rate of Al extraction and the Fe, Na₂O content in the solid residue, the experiments were done according to the method described in section 2.5. Using the experimental results, a neural network model of the process was obtained. Figure 6 shows the response surfaces of the model constructed by varying time, temperature, and the Na₂O concentration of the solution in the leaching process of the BR.

The major effect on Al extraction degree (Figure 6a and Figure 6b) is caused by leaching time and temperature. Figure 6a shows that before 2.0-3.5 h of leaching, the Al extraction degree is increasing. After 3.5 h at 100 °C and 3 h at 110 °C, Al begins to precipitate in the form of desilication product (DSP), which can be confirmed by the increased content of Na₂O in the solid residue after 4 h of leaching (Figure 6e and Figure 6f). After 4 hours of leaching, the iron content in the solid residue decreases because of the DSP precipitation (Figure 6c and Figure 6d). Nevertheless, a high Al extraction degree was achieved after 2 h at all leaching temperatures – more than 80%. After 2.5 hours of leaching at T = 120 °C, the total Fe content in the solid residue can be increased from 41.7% (Table 1) to 56%. The Na₂O concentration in the solution did not have a significant effect on the Al extraction and the Fe, Na₂O content in the solid residues (Figure 6b, 6d, 6f). Therefore, the effect of Na₂O concentration in the solution on the sands leaching is not shown (Figure 7).



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Figure 6. Neural network response surfaces for effect of time and temperature on the Al extraction from bauxite residue (BR) (a); effect of time and Na₂O concentration on the Al extraction from BR (b); effect of time and temperature on the Fe content in the solid residue (c); effect of time and Na₂O concentration on the Fe content in the solid residue (d); effect of time and temperature on the Na₂O content in the solid residue (e); effect of time and Na₂O concentration on the Na₂O content in the solid residue (f).

According to Figure 6a, the major effect on Al extraction is caused by leaching time. After 2 h of leaching at $T = 120 \text{ °C}$ and $C_{Na_2O} 360 \text{ g L}^{-1}$, the Al extraction degree was > 95%.

After 2-2.5 h of leaching, the solid residue has the highest Fe content (Figure 7b). Therefore, the optimal leaching parameters for Sands and BR can be as follows: $T = 120\text{ }^{\circ}\text{C}$, $C_{\text{Na}_2\text{O}} = 360\text{ g L}^{-1}$, and leaching time = 2.5 h. The content of Fe in the sands solid residue at these conditions was about 70%, while the Na_2O content was less than 0.25% (Figure 7c) and this product can be used as a pigment-quality magnetite.

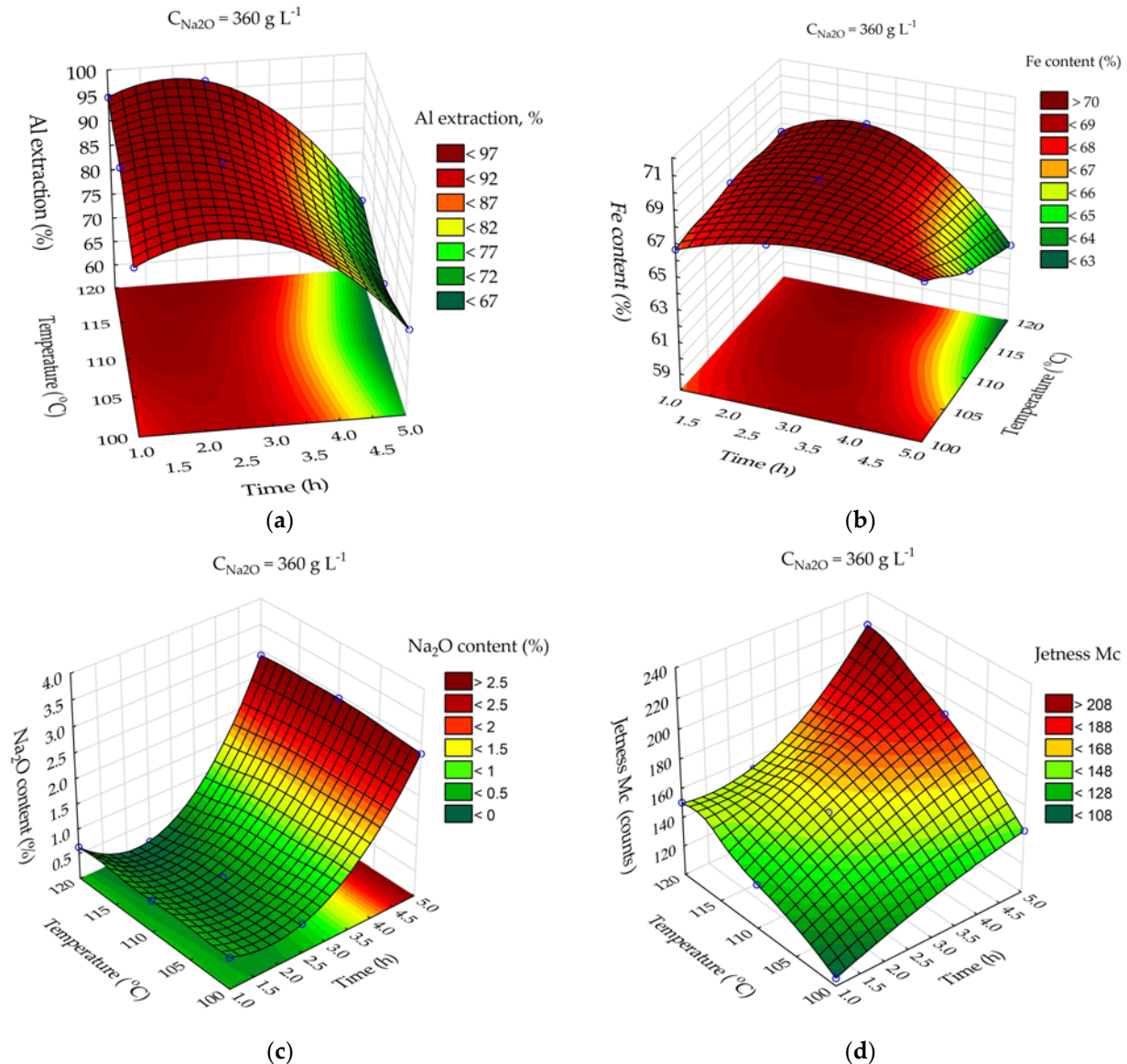


Figure 7. Neural network response surfaces for effect of time and temperature on the Al extraction from sands (a); effect of time and temperature on the Fe content in the sands residue (b); effect of time and temperature on the Na_2O content in the sands residue (c); effect of time and temperature on the jetness (d).

One of the most important characteristics of the pigments is their jetness. Figure 7d illustrates how leaching time and temperature affect the jetness of the magnetite concentrate obtained after the leaching of sands in the presence of Fe^{2+} . According to this data, 5 h of leaching is necessary to achieve a jetness higher than 200. It should be noted, that the jetness of modern pigments can be higher than 300 [43]. Therefore, there is a dilemma in achieving both high jetness and low impurities amounts, because after 5 h of leaching, the Na_2O content increases to 2.5%.

One of the possible solutions to this problem is double leaching, in which Al and Si are removed with a partial transfer of goethite to magnetite at the 1st stage, and the product

is then subjected to a 'curing process' in a pure alkaline solution to obtain the required degree of blackness at the second stage.

3.2. Leaching kinetics

During a fluid-particle heterogenous reaction a solid particle reacts with a liquid with the formation of a solid product. If the solid particle that reacts with a liquid shrinks in size then a shrinking core model can be used to describe this process [44]. According to the model, the reaction has five steps: diffusion through the liquid film, diffusion through the product layer, the surface chemical reaction, liquid or gas product diffusion through the product layer, and product diffusion through the liquid film to the solution. The slowest step of this process is referred to as the rate limiting step. There are three main models of a shrinking core that used to describe the process: Equation (13), which describes diffusion through the liquid film, Equation (14), which describes diffusion through the product layer, and Equation (15), which describes reaction on the surface of the core.

$$X = kt, \quad (13)$$

$$1 - 3(1 - X)^{2/3} + 2(1 - X) = kt, \quad (14)$$

$$1 - (1 - X)^{1/3} = kt. \quad (15)$$

These equations are valid for the leaching of a solid with a uniform particle size distribution [44]. Therefore, Equations (13)-(15) were used to describe the extraction fraction of Al (X) from the different minerals of the raw sands (Figure 8a-8c). Only 2 h leaching time was used in the kinetics calculation because after this time the fresh DSP began to precipitate (Figure 6). The gibbsite that was not leached during the Bayer process and the Al that can be included in the Al-goethite were denoted as a "solid matrix" and the Al that contained in the DSP was denoted as a "DSP". The results of fitting the experimental data in Equations (12)-(14) are shown in Table 4, where the correlation coefficient values (R^2) and the corresponding leaching rate constants (k) for all plots in Figure 8a-8c are listed. The intraparticle diffusion model (Equation (13)) showed the highest convergence with the extraction of Al from the DSP. Because the DSP has a low leaching efficiency, it can be assumed that the product layer consists of silica-containing minerals that are insoluble at these leaching conditions.

During "Solid matrix" phase leaching, there is no new solid product, but the solid matrix of iron minerals that includes Al-goethite can play a role of a reaction solid product. However, the model of the surface chemical reaction (Equation 14) was best to fit the data of Al extraction from the "solid matrix" phase. The Arrhenius plots of $\ln k$ versus $1000/T$ were constructed after the rate constants for both processes were calculated. The activation energy for Al extraction from "DSP" phase was 30.9 kJ/mol, and for Al extraction from "solid matrix" phase it was 36.7 kJ/mol. The activation energy greater than 40 kJ/mol is typical for a leaching process limited by the surface reaction [45]. Therefore, it confirms that the extraction of Al from a "solid matrix" phase is more temperature dependent and can be limited by the surface reaction.

Figure 8 shows that the Al extraction degree from the solid matrix of iron containing minerals is high (almost 100 %) using leaching with the transformation of goethite (hematite) to magnetite. This can be explained by the fact that goethite should be dissolved before a new solid phase (magnetite) can form [46]. The Al extraction degree from DSP is relatively low because sodalite is insoluble in the alkaline media until a new phase is formed, for example hydrogarnet [47]. Moreover, after the concentration of silicon in the solution reaches a certain value, the DSP begins to precipitate again (Figure 6). To enhance the Al extraction degree from DSP, it is necessary to use several leaching cycles or add lime.

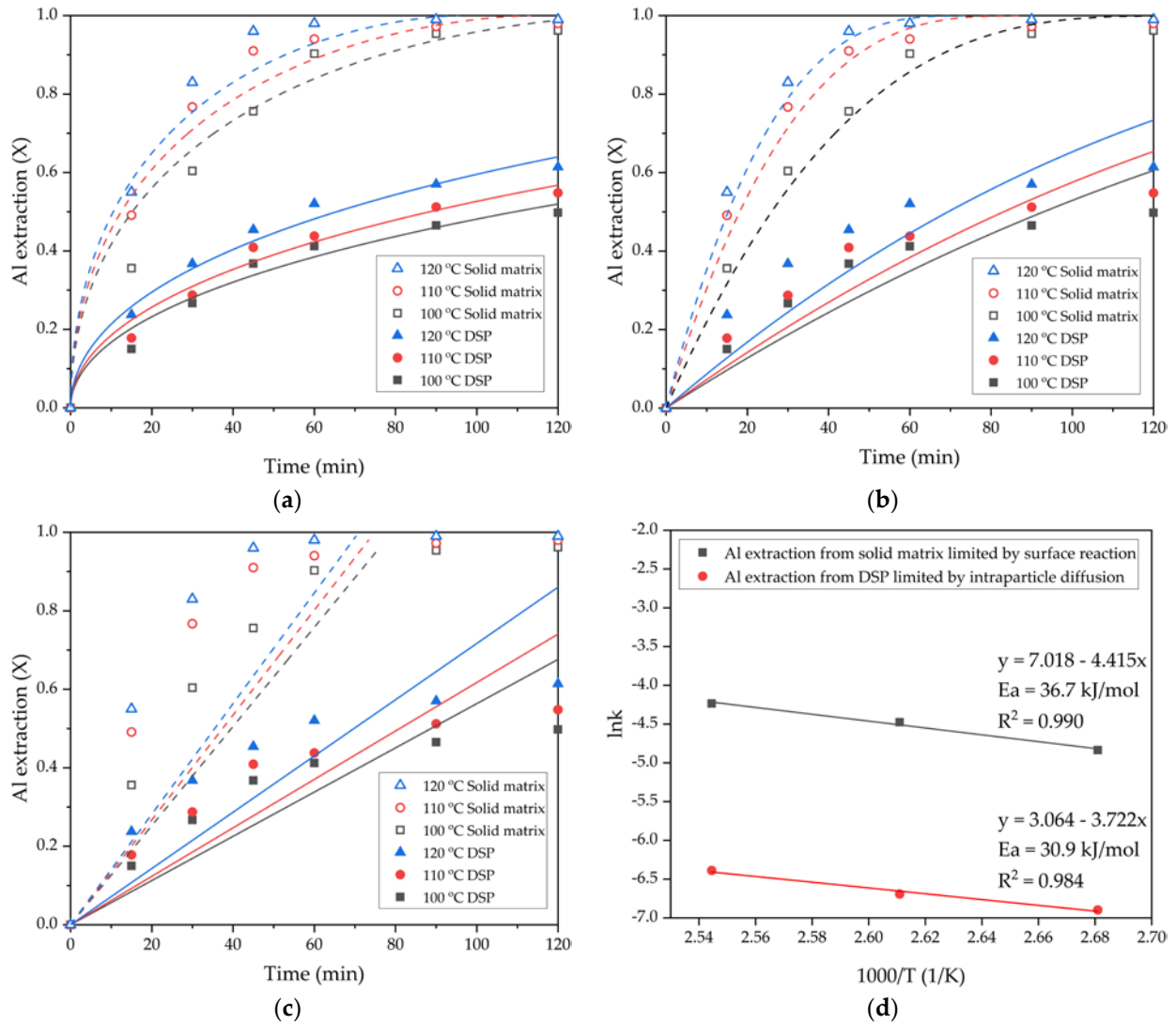


Figure 8. The results of fitting the obtained data for the effect of temperature on the Al extraction from the different phases (points) to Equations (12)-(14) (lines): the diffusion through the product layer (a); the surface chemical reaction (b); the diffusion through the liquid film (c); the apparent rate constants of the Al extraction from the different phases at different temperatures plotted vs. the inverse temperature (Arrhenius plots) (d).

Table 4. The results of the experimental data (figure 7) fitting to the shrinking core models.

Equation	Al containing phase	Temperature, °C	Apparent rate constant k (min ⁻¹)	R ² (%)
Diffusion through the product layer (Figure 7a)	DSP	100	0.0010	0.964
		110	0.0012	0.972
		120	0.0017	0.967
	Solid matrix	100	0.0076	0.904
		110	0.0088	0.865
		120	0.0105	0.800
Surface chemical reaction (Figure 7b)	DSP	100	0.0022	0.851
		110	0.0025	0.861
		120	0.0030	0.830
	Solid matrix	100	0.0079	0.996
		110	0.0114	0.966
		120	0.0145	0.971

Diffusion through the liquid film (Figure 7c)	DSP	100	0.0044	0.929
		110	0.0048	0.930
		120	0.0055	0.912
	Solid matrix	100	0.0090	0.870
		110	0.0093	0.852
		120	0.0095	0.832

3.3. Solid residues characterisation

3.3.1. Chemical composition of the residues

After 2.5 h of leaching at a ferrous sulfate to bauxite residue ratio of 1:1 and 120 °C in alkaline media, the alumina extraction ratio is 96.27% for Sand and 87.06% for BR. At these conditions, the Fe content in the sands residue and in the BR residues can be increased to 69.55% and 58.31%, respectively. Table 5 shows the chemical compositions of the solid residues obtained after BR and Sands leaching at optimal parameters.

Table 5. The chemical compositions of the solid residues obtained after leaching, wt. % (T = 120 °C, τ = 2.5 h, FeSO₄ to BR ratio of 1:1).

Sample	Fe	Si	Ti	Al	Na	O	Sc	P	S	Other
BR solid residue	58.31	2.06	3.92	1.33	1.45	31.35	0.006	0.007	0.08	1.58
Sands solid residue	69.55	0.34	1.01	0.15	0.24	27.81	0.005	0.009	0.003	0.90

Table 5 shows that the solid residue from leaching of sands contains low amounts of impurities: Na and Al are less than 0.25%. The content of Si and Ti in this residue is also smaller than in traditional bauxite residues. The higher contents of Na, Ti, Si and Al in the solid residue obtained after BR leaching makes it difficult to use this product as a pigment. The high Na content will also hinder its use for iron production. Therefore, preliminary leaching with acid or lime causticisation should be used to further improve the valorization of the BR residues. The P and S content in these products is lower than 0.1% (Table 5), which would be beneficial for iron production.

To evaluate why there is such a difference in the results obtained using BR and Sands, the XRD patterns (Figure 3) of the solid residues were investigated. The higher amount of hematite (Al-hematite) in BR (Figure 3) results in the incompleteness of the process (the formation of magnetite from goethite is a faster process from a thermodynamic viewpoint) [48]. Furthermore, there is a higher amount of sodalite (DSP) in BR that will not fully dissolve in the alkaline media without lime addition, which leads to the higher Na content in the residue. From the XRD patterns, it can be confirmed that the conversion of goethite and hematite to magnetite is more effective for sands. To support this hypothesis, a SEM-EDX and the Mössbauer analysis of the solid residue was conducted.

According to elemental mapping of the solid residues (Figure 9a and 9b), it is obvious that the BR solid residue has high amount of Al, Si and lesser amount of Fe. This confirms previous observations. Furthermore, the particle size of the Sands residue is smaller (see Figures 9c and 9d), and there can be seen particles of the raw BR in Figure 9e.

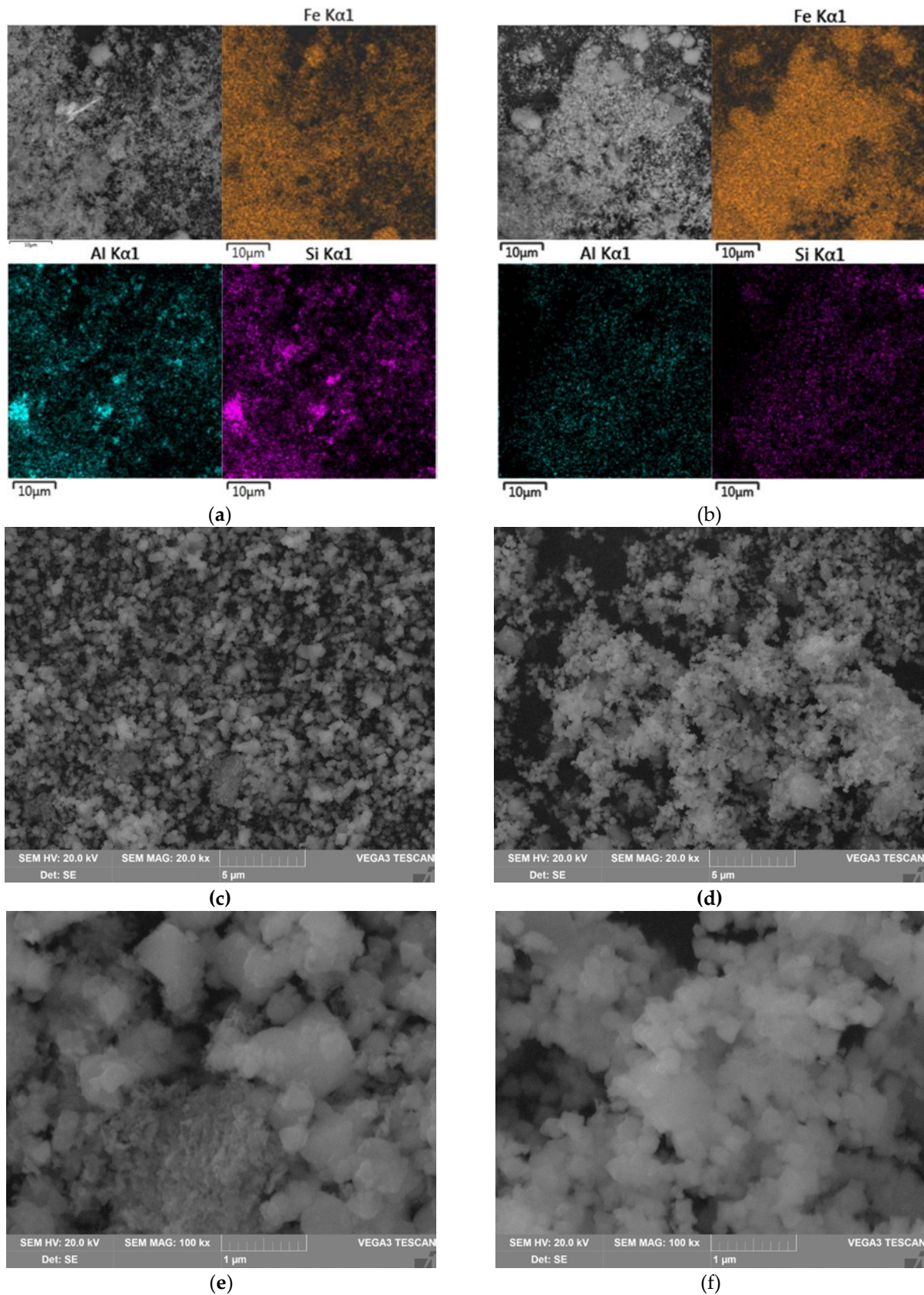


Figure 9. Mapping and morphology of BR and Sands residue surface using SEM-EDX analysis: BSE image and mapping of the BR residue surface (a); BSE image and mapping of the Sands residue surface (b); SEM image of the BR residue surface at 20 kx magnitude (c); SEM image of the Sands residue surface at 20 kx magnitude (d); SEM image of the BR residue surface at 100 kx magnitude (e); SEM image of the Sands residue surface at 100 kx magnitude (f).

3.3.2. Mössbauer analysis

At room temperature, the Mössbauer spectra of the BR and Sands residue samples contain six asymmetric lines of different widths and intensities, some of which split into two in the region of negative velocities (Figure 10). At the same time, significant extended absorption is observed in the central part of the spectrum of the BR residue sample. When the samples are cooled to the boiling point of nitrogen, the lines usually narrow, but their splitting and the presence of shoulders indicate that the iron atoms have several crystallographic positions that differ significantly from each other. Indeed, experimental spectra can only be satisfactorily described using at least four symmetrical sextets (and one doublet for high-temperature spectra) (Table 6).

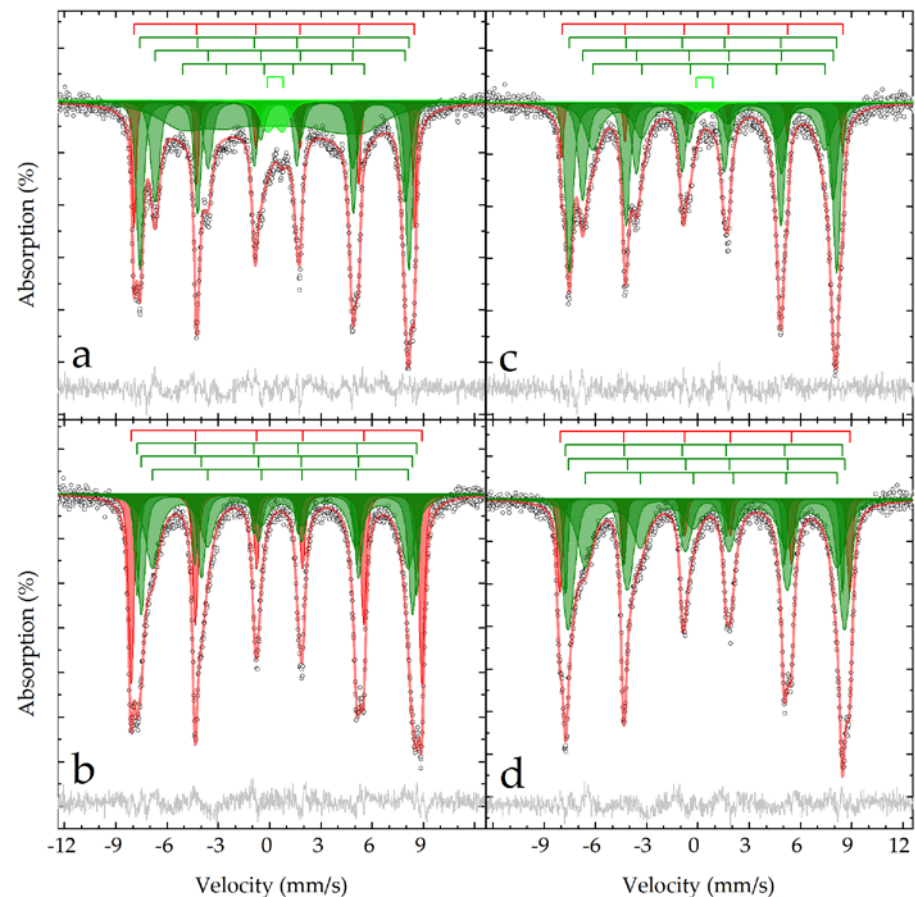


Figure 10. Mössbauer spectra obtained at 296 K (a, c) and 77.6 K (b, d) of the BR residue (a, b) and Sands residue (c, d) samples and their model description according to Table 6.

The outer sextet with rather narrow lines belongs to the Al-hematite described above for the raw materials. Moreover, if for the raw materials the proportion of Al-hematite in the BR sample was almost one and a half times higher (Table 2), then after treatment the proportion of Al-hematite in the BR residue sample is two times higher compared to the Sands residue sample (Table 6). Similarly, to the procedure described above, using equation (12), one can estimate the compositions of Al-hematite as $\alpha\text{-(Fe}_{0.96}\text{Al}_{0.04}\text{)}_2\text{O}_3$ and $\alpha\text{-(Fe}_{0.94}\text{Al}_{0.06}\text{)}_2\text{O}_3$ in the corresponding samples.

The remaining three sextets correspond to iron atoms in different crystallographic positions for nonstoichiometric magnetite of the composition $\text{Fe}_{3-\delta}\text{O}_4 \equiv (\text{Fe}^{3+})_A(\text{Fe}^{2+}_{1-3\delta}\text{Fe}^{3+}_{1+2\delta})_B\text{O}_4$ [49,50]. By analyzing the areas and isomeric shifts of the subspectra related to iron atoms in different crystallographic sites of magnetite, it is possible to estimate the value of the magnetite nonstoichiometric parameter δ , using the Equation (16) [51]:

$$\delta = \frac{\{\sum(\delta_2 - 3\delta_1 + 2\delta_3) * S_i + (\delta_2 - \delta_3) \sum S_j\}}{\{\sum(3\delta_2 - \delta_1 - 2\delta_3) * S_i + 3(\delta_2 - \delta_3) \sum S_j\}}, \quad (16)$$

where S_i is the relative area of the subspectrum with isomeric shift δ_i related to iron atoms in the B-site, S_j is the relative area of the remaining subspectra, δ_2 and δ_3 are isomeric shifts of iron atoms (+2) and (+3) in an octahedral oxygen environment for a given temperature (here $\delta_2 = 1.16 \pm 0.06$ and 1.33 ± 0.09 mm/s for 296 and 78 K, respectively, and $\delta_3 = 0.37 \pm 0.04$ and 0.49 ± 0.04 mm/s for 296 and 78 K, respectively [41]). The results obtained for different temperatures are in good agreement with each other and make it possible to describe the composition of the samples as $\text{Fe}_{2.75}\text{O}_4$ and $\text{Fe}_{2.85}\text{O}_4$ for BR residue and Sands residue, respectively, so the first sample is more oxidized.

The hyperfine parameters of the minor doublet observed only at room temperature correspond to Fe^{3+} ions in an octahedral oxygen environment [41].

Table 6. The subspectrum parameters describing experimental Mössbauer spectra obtained at different temperatures for BR residue and Sands residue samples.

Temperature, K	Sample	Phase	N_0	BR residue					Sands residue						
				δ	$\epsilon (\Delta=2\epsilon)$	Γ_{exp}	H_{eff}	S	δ	$\epsilon (\Delta=2\epsilon)$	Γ_{exp}	H_{eff}	S		
				mm/s		kOe	%		mm/s		kOe	%			
296	$\alpha\text{-(Fe}_{1-x}\text{Al}_x)_2\text{O}_3$	1		0.38 ± 0.01	-0.10 ± 0.01	0.28 ± 0.01	508.7 ± 0.2	12.4 ± 0.6		0.38 ± 0.01	-0.11 ± 0.01	0.24 ± 0.01	509.8 ± 0.3	6.1 ± 0.4	
		2		0.32 ± 0.01	-0.03 ± 0.01	0.49 ± 0.01	488.4 ± 0.2	29 ± 1		0.32 ± 0.01	-0.02 ± 0.01	0.52 ± 0.01	485.3 ± 0.2	40.3 ± 0.8	
	$\text{Fe}_{3-x}\text{O}_4$	3		0.63 ± 0.01	-0.01 ± 0.01	0.62 ± 0.01	454.0 ± 0.2	22.4 ± 0.7	0.15	0.62 ± 0.01	-0.03 ± 0.01	0.61 ± 0.02	454.4 ± 0.4	27 ± 2	0.25
		4		0.38 ± 0.01	-0.16 ± 0.03	3.5 ± 0.2	330 ± 4	32 ± 2		0.65 ± 0.01	-0.00 ± 0.01	1.06 ± 0.04	422 ± 2	25 ± 2	
		$\text{Fe}^{3+}_{\text{Oh}}$	5		0.32 ± 0.02	(0.94 ± 0.03)	0.89 ± 0.07		4.3 ± 0.3		0.38 ± 0.01	(0.94 ± 0.01)	0.9 ± 0.2		1.6 ± 0.2
77.6	$\alpha\text{-(Fe}_{1-x}\text{Al}_x)_2\text{O}_3$	1		0.50 ± 0.01	-0.09 ± 0.01	0.37 ± 0.01	527.3 ± 0.2	27.4 ± 0.8		0.49 ± 0.01	-0.08 ± 0.01	0.36 ± 0.01	526.0 ± 0.3	12.6 ± 0.9	
		2		0.40 ± 0.01	0.03 ± 0.01	0.37 ± 0.01	507.7 ± 0.4	15 ± 1		0.37 ± 0.01	-0.01 ± 0.01	0.30 ± 0.01	502.6 ± 0.2	10.5 ± 0.8	
	$\text{Fe}_{3-x}\text{O}_4$	3		0.52 ± 0.01	-0.09 ± 0.01	0.55 ± 0.02	492.5 ± 0.5	26 ± 2	0.16	0.52 ± 0.01	-0.03 ± 0.01	0.87 ± 0.02	501.6 ± 0.7	42 ± 1	0.24
		4		0.67 ± 0.01	-0.04 ± 0.01	1.00 ± 0.03	464.2 ± 0.8	32 ± 1		0.84 ± 0.01	-0.07 ± 0.01	1.12 ± 0.03	457.2 ± 0.9	35 ± 1	

δ - Isomer shift; $\epsilon (\Delta)$ - Quadrupole shift (splitting); Γ_{exp} - Line width; H_{eff} - Hyperfine magnetic field; S - Relative area of a subspectrum N_0 , " δ in $\text{Fe}_{3-x}\text{O}_4$ " - magnetite nonstoichiometric parameter [51].

3.3.3. Magnetization

Figure 10 shows the magnetization measurements of the samples. Both raw samples appear to be magnetically inert. The residues magnetization reaches a saturation value (Ms) of 45–60 emu g^{-1} at a magnetic field of 10 kOe. These values are lower than those of pure synthetic magnetite according to the literature (70–90 emu g^{-1}) [55]. However, the values are comparable with those determined for magnetite concentrates obtained from BRs [25]. As shown in Figure 10, the magnetization of both the BR residue and sand residue increases with increasing leaching duration. Also, the Ms of the BR residues is lower at all leaching durations. This can be explained by the higher level of impurities in the case of the BR residues.

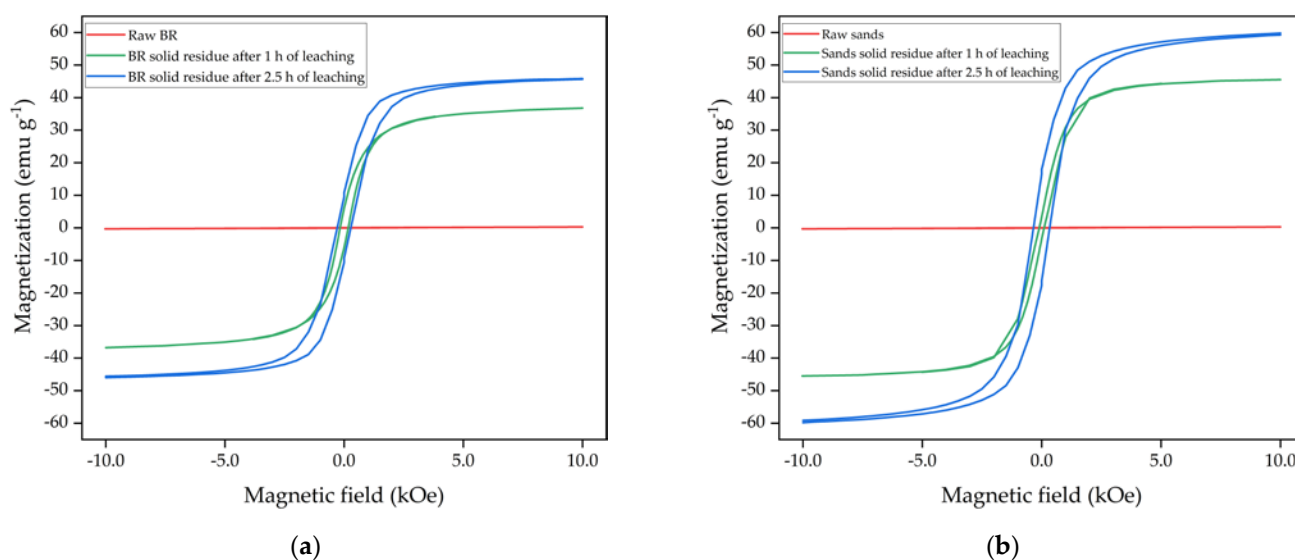


Figure 10. Magnetization curves for the BR and the BR solid residues obtained after BR leaching at 110 °C, $C_{\text{Na}_2\text{O}}$ 360 g L⁻¹ during 1h and 2.5 h (a); the sands and the sands solid residues obtained after BR leaching at 110 °C, $C_{\text{Na}_2\text{O}}$ 360 g L⁻¹ during 1h and 2.5 h (b).

4. Conclusions

In this study, the novel method of treating high-iron bauxite residues from the Friguia alumina refinery using low-temperature leaching in the presence of Fe^{2+} was investigated.

According to the XRD, TG-DTA and Mössbauer analysis, aluminum in this type of BR is presented by Al-hematite and Al-goethite, that are insoluble in the common Bayer leaching. The leaching in the presence of Fe^{2+} accelerates Al extraction from Al-hematite and Al-goethite. This effect is caused by the magnetization of magnetite and goethite after they are dissolved in a concentrated alkaline solution. The analysis of the Al extraction degree, Fe and Na_2O content in residue obtained by alkaline leaching in the presence of Fe^{2+} was performed using ANNs and machine learning. The optimal leaching parameters were found to be $T = 120$ °C, $L:S = 10$, and $\tau = 2.5$ h, $C_{\text{Na}_2\text{O}} = 360$ g L⁻¹. At these parameters, the alumina extraction degree reached 96.27% for sand and 86.07% for bauxite residue. The grade of iron (total iron in the form of iron element) in the residue can be increased to 69.55% for Sands and 58.31% for BR.

The extremely low impurities content in the solid residue obtained after the Sands leaching and the high blackness make it a good alternative for pigment production. The difference in extraction degree and total Fe content in the solid residues obtained from Sands and BR can be attributed to the higher amount of Al-hematite and sodalite in BR. Sc acid leaching or lime causticization can be used to reduce the amount of sodalite in BR before it is valorized.

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