Monosodium Glutamate as Selective Lixiviant for Alkaline Leaching of Zinc and Copper from Electric Arc Furnace Dust

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Abstract: The efficacy of monosodium glutamate (MSG) as a lixiviant for the selective and sustainable leaching of zinc and copper from electric arc furnace dust was tested. Batch leaching studies and XRD, XRF and SEM-EDS characterization confirmed the high leaching efficiency of zinc (reaching 99%) and copper (reaching 86%) leaving behind Fe, Al, Ca and Mg in the leaching residue. The separation factor (concentration ratio in pregnant leach solution) between zinc vs. other elements, and copper vs. other elements in the optimum condition could reach 11,700 and 250 times, respectively. The optimum conditions for the leaching scheme were pH 9, MSG concentration 1 M and pulp density 50 g/L. Kinetic studies (leaching time and temperature) revealed that the saturation value of leaching efficiency was attained within 2 h for zinc and 4 h for copper. Modeling of the kinetic experimental data indicated that the role of temperature on the leaching process was minor. The study also demonstrated the possibility of MSG recycling from pregnant leach solutions by precipitation as glutamic acid (>90% recovery).

Keywords: electric arc furnace dust; monosodium glutamate; leaching; zinc; copper

1. Introduction

The demand for base metals such as zinc (Zn) and copper (Cu) has increased in recent years. Natural (primary) resources of both metals are dominated by sulfide ores, which unfortunately have declined both in quantity and grade causing discrepancies between supply and demand [1]. This in turn encourages the exploration of new resources, which not only includes primary resources but also secondary ones through recycling. Aside from helping address the resource depletion problem, the exploitation of secondary resources also mitigates environmental and resource sustainability problems. One of the secondary resources of base metal, which holds potential for further processing, is electric arc furnace (EAF) dust. The dust is a waste of steel making and is classified as hazardous [2]. The amount of dust produced during steel making is significant, where 11–20 kg of dust is generated for each ton of steel produced [3].

In addition to exploring new resources, extraction is also progressing toward green technology, which prioritizes the principles of efficiency, sustainability, safety and environmental impact. The EAF dust not only contains Zn and Cu as target elements but also others, e.g., lead (Pb), chromium (Cr),
iron (Fe), aluminum (Al), calcium (Ca) and magnesium (Mg). As Pb and Cr (VI) represent pollutants, their release into the environment during the extraction of Zn and Cu must be minimized and as such, a comprehensive study is required to formulate applicable technology to maximize benefits from EAF dust and minimize the impact of pollutants (heavy metals) to the environment.

One of the most widely applied methods to recover Zn and Cu from EAF dust is hydrometallurgy, especially in the case where Zn in EAF dust exists as zinc oxide. Other common Zn phases, i.e., zinc ferrite requires pre-treatment (e.g. fusion) [4,5]. The hydrometallurgical approach for Zn and Cu recovery involves a leaching process using chemical lixiviants to dissolve these metals in aqueous solution. Dissolved Zn and Cu in the pregnant leach solution can then be further separated and purified using solvent extraction [6,7], ion exchange [8] or electrochemical methods [9,10]. Several leaching schemes that use various lixiviants have been previously proposed in order to recover Zn and/or Cu from EAF dust at ambient and elevated temperatures, respectively. Halli et al., 2017 [11] had screened potential lixiviants including nitric acid, citric acid, sodium hydroxide and ethanol for metal recovery from EAF dust. Proposed lixiviants by other researchers include sulfuric acid [12–14], hydrochloric acid [15], oxalic acid [16], citric acid [17], sodium hydroxide [18–20], sodium carbonate/sodium bicarbonate [21], ammonia [22], iminodiacetic acid (IDA) [23] and nitrilotriacetic acid (NTA) [24].

In general, the leaching schemes proposed in the literature can be classified into four categories:

1. Acidic leaching using strong acid, e.g., sulfuric acid, hydrochloric acid.
2. Acidic leaching using organic lixiviants, e.g., oxalic acid, citric acid.
3. Alkaline leaching using bases, e.g., sodium hydroxide, sodium carbonate.
4. Alkaline leaching using organic lixiviants, e.g., nitrilotriacetic acid (NTA).

The first and second schemes are advantageous in terms of recovery efficiency but both lack selectivity. Other elements, especially Fe and Ca which would cause fouling in further separation and purification steps, and Pb or Cr which are toxic, are also extracted [25]. The third scheme possesses an advantage in terms of selectivity but the extraction efficiency is low, requiring very low pulp density ratio, high lixiviant concentration and high leaching temperature to attain satisfying recovery. In addition, the caustic and corrosive nature of lixiviant requires special handling and equipment during operation and also poses a risk to the environment. The same is true of schemes using strong acid as lixiviant to certain extent, especially leaching using high concentration (concentrated) mineral acid, e.g., Halli et al., 2017 [11].

Considering these factors, the fourth scheme presented is preferable, due to its relatively selective recovery of Zn and Cu, higher leaching efficiency, safer handling in general compared to lixiviants in the other schemes and possibility to recover the lixiviant such as NTA. Although the leaching of base metal in this scheme is considered as selective, previous studies by Yang et al. (2016) [24] revealed relatively high extraction of iron, while the recovery of other elements such as Ca, Mg and Al during leaching was yet to be addressed, where these elements play an important role in further separation and purification steps. The lixiviants in the fourth category rely on their ability to form complexes with metal ions. Aminocarboxylic acid ligands, e.g., IDA, NTA or ethylenediaminetetraacetic acid (EDTA) possess relatively strong yet homogenous binding capacity to the metal ions, not only target elements such as Cu and Zn, but also matrix elements i.e. Fe, Al, Mg and Ca.

In order to partition the matrix elements into the solid phase and the valuable elements into the aqueous phase, it is suggested to use a ligand which complexes preferentially to Cu and Zn and shows weaker affinity to Ca, Mg, Fe and Al. Lixiviants such as amino acids broadly meet these criteria. Glycine has been extensively studied in copper [26] and gold [27] alkaline leaching, but to the best of our knowledge, no study on the efficacy of amino acid on Zn recovery has been carried out so far. Aside from glycine, another amino acid with potential to be developed as lixiviant is glutamic acid. In this research an amino acid sodium salt, i.e., monosodium glutamate (MSG) is proposed as a novel lixiviant. The advantages of MSG as a lixiviant include its wide availability and low price, and low risk to environment (biodegradable). MSG in alkaline conditions could serve as a powerful ligand to
bind several transition metals, e.g., Cu, which in turn increase the extraction efficiency. The complex formation of these metals with MSG have been reported and exploited in chemical analysis [28,29] and our preliminary research also confirmed its efficacy to extract Zn [30]. Another advantage of using MSG is its reusability, since MSG can be recovered from the pregnant leach solution as glutamic acid precipitate by acidification of pregnant leach solution (PLS). Considering these possible advantages of using MSG, a study to confirm its efficacy in Zn and Cu extraction from EAF dust is warranted. The effect of leaching parameters on the leaching efficiency such as pH, lixiviant concentration, pulp density and kinetics, including selectivity toward other elements are investigated herein.

2. Species Distribution Modeling for Leaching Efficiency and Selectivity Prediction

The leaching efficiency and selectivity are hypothetically controlled by the distribution of species in the aqueous phase. Factors controlling this distribution include pH and concentration of metal ions and ligands/lixiviants involved in the system. The effect of pH on the recovery and selectivity could be predicted using a species distribution diagram, which was constructed using Visual MINTEQ 3.1 (KTH Royal Institute of Technology, Sweden). [31]. Figure 1 shows the species distribution of Zn and Cu in glutamate-H2O system as a function of pH.

![Species distribution diagram](image1.png)

*Figure 1. Species distribution of Zn (a) and Cu (b) in glutamate-H2O system as function of pH. Glutamate concentration 1 M, Zn\(^{2+}\) 300 mM and Cu\(^{2+}\) 8 mM.*

Based on Figure 1, in acidic conditions the species are dominated by free Zn\(^{2+}\) and Cu\(^{2+}\) ions. Glutamate as chelate effectively binds Zn and Cu from weakly acidic to alkaline conditions, and prevents the precipitation of hydroxides at strong alkaline pH. The figure also shows that Zn starts to hydrolyze at pH 12, while Cu at pH 13. The program was also used to model the species distribution according to pH for other elements (Supplementary file: Figure S1), i.e., Fe, Mg, Ca and Al. In general Fe and Al start to hydrolyze at weakly acidic pH (±4), while Mg and Ca tend to hydrolyze at medium alkaline conditions (pH 12).

Aside from species distribution, leaching selectivity can be predicted using the complex formation constant \((K_f)\) (Table 1). The constant was calculated for each complex, which might exist in the solution phase. Table 1 also lists the formation constants of metal ions with EDTA, NTA and glycinate as comparisons to assess the selectivity of organic compounds as lixiviants. If a lixiviant is expected to selectively bind one metal ion \((\alpha)\) and not another metal ion \((\beta)\), then the value of \(K_{fa}/K_{fb}\) has to be higher than 10\(^6\) \((\log K_{fa} - \log K_{fb} \geq 6)\) [32]. The value of complex formation constants for glutamate, which were calculated based on the model generated by Visual Minteq in Table 1, indicates the possibility of separation of Zn and Cu from Mg and Ca, and also Zn from Cu using glutamate, and demonstrates better selectivity over the other chelators such as EDTA, NTA and glycine (amino acid).
The weaker complex formation constants for Fe-glutamate compared to Fe-EDTA and Fe-NTA indicate the possibility of separation of Zn and Cu from Fe through Fe hydrolysis in alkaline condition.

**Table 1.** Log complex formation constant of metal ions with three organic compounds as chelator/lixiviants.

| Ions    | log $K_f$ EDTA [33] | log $K_f$ NTA [34] | log $K_f$ Glycinate [35] | log $K_f$ Glutamate (This Study) |
|---------|---------------------|--------------------|--------------------------|---------------------------------|
| Zn$^{2+}$ | 16.4               | 10.7               | 5.5                      | 8.9                             |
| Cu$^{2+}$ | 18.4               | 12.7               | 8.3                      | 14.9                            |
| Fe$^{3+}$ | 24.2               | 24.3               | 10.3                     | 11.8                            |
| Mg$^{2+}$ | 8.7                | 8.2                | 3.5                      | 1.8                             |
| Ca$^{2+}$ | 10.6               | 7.0                | 1.4                      | 1.1                             |

3. **Experimental**

3.1. **Materials and Instrumentation**

EAF dust samples were kindly supplied by local smelters around Jakarta and Banten Province, Indonesia. A sieving test showed that more than 90% of dust passed through a 270 mesh (53 µm), and this size fraction was used in further leaching tests. All leaching experiment and characterization studies were performed in the Research Unit for Mineral Technology—Indonesian Institute of Sciences, Bandar Lampung, Indonesia, or otherwise stated.

XRF characterization (Panalytical X'Pert 3 Powder with Omnian Standard) was carried out to determine the major chemical components in EAF dust (Table 2), and shows the most dominant constituents to be zinc and aluminum. The mineralogical phases were determined using X-ray powder diffraction (Panalytical, Expert3 Powder). Quantitative determination of metal contents in EAF dust (Table 3) and pregnant leach solution after leaching for recovery and separation factor calculation were carried out using atomic absorption spectrophotometry (AAS, Shimadzu AA7000, Japan) and Inductively Coupled Plasma-Optical Emission Spectrometry (ICP-OES, Analytik Jena, Plasma Quant 9000 Elite, Germany). SEM-EDS characterization to investigate the change of morphology and elemental distribution on the surface of EAF dust grain before and after leaching was performed using a SEM (Hitachi SUM 3500, Japan) at the Research Unit for Natural Product Technology—Indonesian Institute of Sciences, Yogyakarta, Indonesia. The advanced mineral identification and characterization system (AMICS) analysis to determine and map the mineral phases was carried out by Eagle Engineering, Butte, Montana, USA.

Chemicals such as sodium hydroxide, sulfuric acid, nitric acid, hydrochloric acid, and standard solutions were obtained from Merck, Darmstadt all in analytical grade, while monosodium glutamate was produced by PT Ajinomoto Indonesia (99% purity) and used as received. The solution pH was adjusted using dilute sulfuric acid or sodium hydroxide and monitored using a pH meter (Oakton 45, Vernon Hills, IL, USA), while deionized water (MilliQ) was used throughout the experiment.

**Table 2.** Major chemical composition of electric arc furnace (EAF) dust.

| Component | MgO | Al$_2$O$_3$ | SiO$_2$ | SO$_3$ | K$_2$O | CaO | TiO$_2$ | MnO | Fe$_2$O$_3$ | NiO | CuO | ZnO | Total |
|-----------|-----|-------------|--------|--------|-------|-----|---------|-----|-------------|-----|-----|-----|-------|
| wt.%      | 0.59| 36.0        | 2.22   | 0.34   | 0.12  | 2.23| 0.18    | 0.87| 2.72        | 0.60| 1.29| 51.8| 99.2  |

**Table 3.** Chemical composition of important base metals in EAF dust (after aqua regia digestion and AAS/ICP-OES determination).

| Element | Cr        | Cu        | Fe        | Ni        | Pb        | Zn         |
|---------|-----------|-----------|-----------|-----------|-----------|------------|
| wt.%    | 0.016 ± 0.002 | 0.898 ± 0.021 | 1.432 ± 0.029 | 0.522 ± 0.008 | 0.075 ± 0.016 | 38.652 ± 0.428 |
3.2. Leaching Procedure

Leaching studies were carried out using a batch method. In general, 1 g of EAF dust was mixed with 20 mL lixiviant in 250 mL sealed flask. The mixture was homogenized using an orbital shaker at 200 rpm. After the leaching was completed, the supernatant solution was separated using centrifugation and filtration (Whatman 42). The metal concentration was determined using AAS or ICP-OES. Metal recovery \( R \) was calculated using Equation (1). To evaluate the selectivity of the leaching scheme, a separation factor (SF) between Zn or Cu and other metals was evaluated using Equation (2). All leaching data were obtained in duplicates.

\[
R = \frac{C_E \times V}{C_o \times m} \times 100\% \tag{1}
\]

where:

\( C_E \) Zn or Cu concentration in supernatant solution (mg/L)
\( C_o \) Zn or Cu content in EAF dust (mg/g)
\( m \) mass of EAF dust used in leaching (g)
\( V \) leaching agent volume (L)

\[
SF = \frac{C_Z}{C_M} \tag{2}
\]

where:

\( C_Z \), Zn or Cu concentration in pregnant leach solution (mg/L).
\( C_M \), other metal concentration in pregnant leach solution (mg/L).

4. Results and Discussion

4.1. Characterization

EAF dust characterization using XRD showed the mineral phase composition was dominated by zinc oxide (ZnO). XRD of the dust residue obtained after leaching (leaching condition pH 9, MSG concentration 1 M, pulp density 50 g/L and 12-hour homogenization) revealed the phases were dominated by spinel (Mg-Al oxide), calcium aluminum oxide and zinc oxide phases (Figure 2). These elements such as Al, Mg, Ca and Fe were found to be relatively retained in the solid phase during the leaching process. The XRD characterization results are supported by AMICS analysis results (Supplementary file: Tables S1 and S2 and Figure S2), which revealed the major Zn phase to be ZnO and gahnite (ZnAl_2O_4) in original material.
3.2. Leaching Procedure

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\[
\text{Metal recovery} = \frac{\text{Metal concentration in supernatant solution}}{\text{Metal concentration in pregnant leach solution}} \times 100%
\]

\[
SF = \frac{C_{\text{Zn or Cu}} - C_{\text{other metals}}}{C_{\text{other metals}}}
\]

where:
- \( C_{\text{Zn or Cu}} \): Zn or Cu concentration in pregnant leach solution (mg/L).
- \( C_{\text{other metals}} \): other metal concentration in pregnant leach solution (mg/L).
- \( C_{\text{Zn or Cu}} \): Zn or Cu concentration in supernatant solution (mg/L).

Table 4.

| Element | Before leaching (wt %) | After leaching (wt %) |
|---------|------------------------|-----------------------|
| Zn      | 24.08                  | 35.53                 |
| Al      | 1.62                   | 0.57                  |
| Mg      | 15.19                  | 20.42                 |
| Fe      | 1.06                   | 0.90                  |
| Ca      | 0.57                   | 1.00                  |
| Si      | 56.95                  | 6.99                  |

Table 4. Qualitative mapping of elements by EDS on grain surface before and after leaching.

The change in elemental distribution of EAF dust before and after leaching was also explored by SEM-EDS characterization (Figure 3). Surface morphology of EAF dust changed from porous and grainy to relatively smooth. Intensity decrease of Zn and Cu peaks on the EDS profile after leaching confirmed the efficacy of MSG as lixiviant in EAF dust leaching. Qualitative mapping on the surface reveals that elements such as Al, Fe and Ca were relatively retained and even enriched after leaching (Table 4), indicating the selective nature of the leaching using MSG.

\[ \text{Figure 2. Phases present in EAF dust (a) before and (b) after leaching.} \]

\[ \text{Figure 3. Morphology (a) before and (b) after leaching, including EDS intensities of elements based on mapping on the respective field (c) before and (d) after leaching.} \]
Table 4. Qualitative mapping of elements by EDS on grain surface before and after leaching.

| Elements | O K | Mg K | Al K | Si K | Ca K | Fe K | Zn K |
|----------|-----|------|------|------|------|------|------|
| Before leaching (wt %) | 24.08 | 1.62 | 15.19 | 1.06 | 0.57 | 0.37 | 56.95 |
| After leaching (wt %) | 35.53 | 0.57 | 20.42 | 0.90 | 1.00 | 1.30 | 6.99 |

4.2. Effect of pH

The effect of pH on the Zn and Cu leaching efficiency and selective recovery was studied between 6 and 11. The constant variables included MSG concentration 1 M, pulp density 50 g/L and an agitation time of 12 h at room temperature. pH ranges between 6–11 were chosen according to the modeling in Section 2 in order to suppress the solubility of other elements (Mg, Ca, Al and Fe). The leaching efficiency and leaching selectivity are shown in Figures 4 and 5, respectively.

Based on Figure 4, the optimum pH to recover Zn and Cu are 9. Low recovery at lower pH was due to the weaker or repulsive interaction between glutamate and Zn or Cu since the glutamate species was dominated as a protonated species, e.g., $\text{H}_3\text{Glu}^+$, $\text{HGlu}^-$. Lower recovery at pH higher than 9 was probably due to the hydrolysis of Zn and Cu, which was lower compared to the modeling results using Visual Minteq ($\text{pH} > 12$ for Zn and $\text{pH} > 13$ for Cu). In the case of other elements, all but Al extraction decreased as pH became alkaline. Increasing recovery of Al in alkaline conditions (0.5% at pH 11) was due to the amphoteric characteristic of Al, which is soluble in excess of alkali as $\text{Al(OH)}_4^-$. Increasing Al recovery at higher pH and decreasing Zn or Cu contributed to the sharp decrease of the selectivity factor of Zn or Cu toward Al (Figure 5). The figure also shows the optimum pH to separate Zn and Cu are 9 and 10, respectively. The highest separation factor is obtained toward iron, which could reach an order of $10^4$ (with Zn) and 400 (with Cu). Lower separation factors were obtained between Zn or Cu and Mg and Ca (reaching an order of 700).

Figure 4. The recovery of (a) Zn, Cu, Mg and Ca and (b) Fe and Al as function of pH.
4.3. Effect of MSG Concentration

The effects of MSG concentration to metal recovery and Zn and Cu selectivity over other elements were studied between 0.1 M and 2 M, with constant variables pH 9, pulp density 50 g/L and 12 h leaching time at room temperature. The results in Figure 6 show that MSG concentration has positive effects on metal recovery. Based on modeling using Visual Minteq (Supplementary file: Figure S3), the minimum MSG concentration to completely solubilize Zn is 0.7 M, and at lower MSG concentration, Zn speciation is dominated by hydroxide precipitate. In the case of Cu, the model shows that MSG concentration has little effect on speciation, which contradicts the results in Figure 6. This is probably due to competition with Zn, which has a 40 times larger concentration. The leaching process at higher MSG concentration increased the recovery of other elements, e.g., Fe, due to increasing soluble species (FeGlu\textsuperscript{+} complex) (Supplementary file: Figure S3). The results in Figure 6 show the optimum recovery of Zn and Cu are attained at 1.2 and 0.7 M, respectively, which at higher MSG concentration did not significantly increase the recovery of both metals. At higher MSG concentration, the recovery of other metals, especially Mg and Ca became substantial, decreasing the selectivity (Figure 7). The optimum MSG concentration based on data in Figure 6 is well supported by results in Figure 7.
4.4. Effect of Pulp Density

To evaluate the effect of pulp density (ratio between lixiviant and EAF dust) on the recovery and selectivity, leaching was carried out at constant variables pH 9, MSG concentration 1 M, 12 h leach at room temperature, while pulp density was varied between 33 and 200 g/L. The results in Figure 8 show the recovery decreased as pulp density increased. Maximum pulp density to obtain optimum recovery of Zn and Cu were 50 and 100 g/L, respectively. In the case of Mg and Ca the optimum pulp density was 66.67 g/L, while in the case of Fe and Al, the recovery was very low (less than 1 percent).

The plot of separation factor value as a function of pulp density (Figure 9) reveals that pulp density has a negative effect on the separation factor. This was probably caused by the increase of solubility of elements especially Mg, Ca, Fe and Al at larger volumes in alkaline conditions. This also indicates the leaching of EAF dust using MSG as lixiviant in alkaline conditions should be performed using continuous methods, e.g., column leaching or heap leaching instead of a batch method, since in the continuous method the volume could be minimized to optimize the selectivity.

Figure 7. Separation factor of (a) Zn: [Zn]/[Fe] (squares), [Zn]/[Al] (diamonds), [Zn]/[Mg] (circles), [Zn]/[Ca] (triangles) and (b) Cu: [Cu]/[Fe] (squares), [Cu]/[Al] (diamonds), [Cu]/[Mg] (circles), [Cu]/[Ca] (triangles) in pregnant leach solution as function of MSG concentration.

Figure 8. The recovery of (a) Zn, Cu, Mg and Ca and (b) Fe and Al as function of pulp density.
The recovery of Zn and Cu were 50 and 100 g/L, respectively. In the case of Cu, the aqueous phase was reductive, which inhibited the oxidation and complexation of Cu by lixiviant. As leaching progressed, the aqueous phase became oxidative, favoring the oxidation and complexation of Cu.

To describe the kinetic process of leaching, three kinetic models were tested to model experimental data: shrinking core model (chemical reaction control), shrinking particle model (film diffusion) and interface transfer and diffusion by Dickinson and Heal (1999) [36]. The fitting results ($R^2$, coefficient of correlation) of each model are listed in Table 5.

**Figure 9.** Separation factor of (a) Zn: [Zn]/[Fe] (squares), [Zn]/[Al] (diamonds), [Zn]/[Mg] (circles), [Zn]/[Ca] (triangles) and (b) Cu: [Cu]/[Fe] (squares), [Cu]/[Al] (diamonds), [Cu]/[Mg] (circles), [Cu]/[Ca] (triangles) in pregnant leach solution as function of pulp density.

**4.5. Kinetic Studies (Effect of Leaching Time and Temperature)**

Kinetic studies were carried out at a time range up to 12 h at three different temperatures (30, 55 and 80 °C), with constant variables pH 9, MSG concentration 1 M and pulp density 50 g/L. The results in Figure 10 show that maximum leaching efficiency was attained within 120 min for Zn and 4 h for Cu at 30 °C. At higher leaching temperature, recovery reached saturation in a shorter period. Figure 10b shows the recovery of Cu is delayed about 1 h. Based on Eh monitoring, in the early stage of leaching the aqueous phase was reductive, which inhibited the oxidation and complexation of Cu by lixiviant. As leaching progressed, the aqueous phase became oxidative, favoring the oxidation and complexation of Cu.

To describe the kinetic process of leaching, three kinetic models were tested to model experimental data: shrinking core model (chemical reaction control), shrinking particle model (film diffusion) and interface transfer and diffusion by Dickinson and Heal (1999) [36]. The fitting results ($R^2$, coefficient of correlation) of each model are listed in Table 5.

**Figure 10.** The effect of leaching time and temperature on recovery of (a) Zn and (b) Cu. In b, the change of Eh in aqueous phase during leaching was obtained at 30 °C (single measurement).
Table 5. Correlation coefficient for experimental data fitting using three kinetic models (SCM, shrinking core model; SPM, shrinking particle model).

| Element | Model                          | Equation                      | Coefficient of Correlation, R² |
|---------|--------------------------------|-------------------------------|--------------------------------|
|         |                                |                               | 30 °C | 55 °C | 80 °C |
| Zn      | SCM (chemical reaction control) | \( k_t = \frac{1}{t} = \frac{1}{1 - R} x^3 \) | 0.768 | 0.715 | 0.710 |
|         | SPM (film diffusion)            | \( k_t = \frac{1}{t} = \frac{1}{1 - R} x^3 \) | 0.717 | 0.635 | 0.622 |
|         | Interface transfer and diffusion | \( k_t = \frac{1}{3} \ln(1 - x) - \frac{1}{1 - R} x - x^3 \) | 0.886 | 0.916 | 0.895 |
| Cu      | SCM (chemical reaction control) | \( k_t = \frac{1}{t} = \frac{1}{1 - R} x^3 \) | 0.880 | 0.847 | 0.912 |
|         | SPM (film diffusion)            | \( k_t = \frac{1}{t} = \frac{1}{1 - R} x^3 \) | 0.860 | 0.802 | 0.885 |
|         | Interface transfer and diffusion | \( k_t = \frac{1}{3} \ln(1 - x) - \frac{1}{1 - R} x - x^3 \) | 0.930 | 0.969 | 0.931 |

\( k \), apparent rate constant (min\(^{-1}\)); \( t \), time (min); \( R \), metal recovery.

Based on the coefficient of the correlation value in Table 5, the interface transfer and diffusion model is the best model to describe the experimental data. Although the shrinking core model and shrinking particle model are more popular models to explain kinetic data, both models are generally based on the assumption of constant lixiviant concentration during leaching [37]. In the leaching process, the concentration of glutamate would decrease due to complex formation. In this study, the model developed by Dickinson and Heal (1999) [36] was adequately applied to analyze the dissolution of Zn and Cu in EAF dust using glutamate.

The value of the apparent rate constant (\( k \)) obtained for each temperature using the interface transfer and diffusion model (Table 6) was used to determine the activation energy (\( E_a \), kJ/mol) using the Arrhenius Equation (3), where \( A \), \( T \), and \( R \) are the frequency factor, temperature, and gas constant, respectively. Relatively low energy activation for both Zn and Cu indicates that the effect of temperature on the leaching process is minor.

\[
 k = Ae^{-\frac{E_a}{RT}} \text{ or } \ln k = \ln A - \frac{E_a}{RT}
\]  

Table 6. Apparent rate constant (\( k \)) attained from fitting of experimental data using interface transfer and diffusion model, which was used to obtain the activation energy (\( E_a \)) using linear regression, including the coefficient of correlation (\( R^2 \)).

| Element | \( k \) (min\(^{-1}\)) | \( E_a \) (kJ/mol) | \( R^2 \) |
|---------|------------------------|-------------------|---------|
|         | \( 30 \^\circ C \) | \( 55 \^\circ C \) | \( 80 \^\circ C \) |
| Zn      | \( 7.3 \times 10^{-4} \) | \( 1.9 \times 10^{-3} \) | \( 5.4 \times 10^{-3} \) | 35.5 | 0.995 |
| Cu      | \( 3.0 \times 10^{-4} \) | \( 1.0 \times 10^{-3} \) | \( 2.2 \times 10^{-3} \) | 35.6 | 0.994 |

4.6. Monosodium Glutamate Recovery

Due to the significant amount of MSG used in leaching and to sustain the leaching scheme, the regeneration of glutamate is required. Glutamate could be recovered from the pregnant leach solution as precipitate by acidifying the solution to the pH between 2–4.5. In this pH range glutamate is precipitated as neutral species glutamic acid, \( \text{H}_2\text{Glu} \) (Supplementary file: Figure S4). The addition of more acid (pH less than 1) caused the formation of soluble cationic species \( \text{H}_3\text{Glu}^+ \). The experiment performed to recover glutamate from the pregnant leach solution showed that the optimum pH was 3, which in this pH more than 90% of the glutamate was recovered as glutamic acid (Figure 11). The recovery efficiency of glutamate was determined gravimetrically.
5. Conclusions

Monosodium glutamate effectively and selectively recovered Zn and Cu from EAF dust, based on batch leaching studies according to leaching efficiency and separation factor values. The optimum conditions for the leaching scheme are pH 9, lixiviant concentration 1 M and pulp density 50 g/L. Studies on the effect of pulp density on the metal recovery and separation factor showed that pulp density correlated negatively to the metal recovery and positively to the separation factor of Zn and Cu to the other elements. This indicates the leaching is better performed using a continuous method, considering the selective recovery of Zn and Cu from other elements. Kinetic studies showed the leaching efficiency reached a saturation value in less than 2 and 4 h for Zn and Cu, respectively. The activation energy obtained from the experimental data modeling revealed the effect of temperature on the leaching process was minor. Further the use of MSG as lixiviant offers a sustainable leaching scheme since MSG is recoverable from the pregnant leach solution and reusable for the next leaching cycle.

Supplementary Materials: The following are available online at http://www.mdpi.com/2075-4701/10/5/644/s1, Figure S1: Species distribution of (a) Al, (b) Fe, (c) Mg and (d) Ca in glutamate-H₂O system as function of pH. Glutamate concentration 1 M, Al^{2+} 350 mM, Fe^{3+} 10 mM, Mg^{2+} 10 mM and Ca^{2+} 20 mM, Table S1: AMICS Mineralogy for Zinc EAF Dust Sample, Figure S2: Identified AMICS Minerals for Zinc EAF Dust Sample, Table S2, AMICS Color Scheme, Figure S3: Species distribution of (a) Zn and (b) Fe in glutamate-H₂O system as function of glutamate concentration at pH 9. Zn^{2+} 300 mM and Fe^{3+} 10 mM, Figure S4: Species distribution of glutamate in glutamate-H₂O system as function of pH. Glutamate concentration 1 M.

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References

1. Ejtemaei, M.; Gharabaghi, M.; Irannajad, M. A review of zinc oxide mineral beneficiation using flotation method. Adv. Colloid Interface Sci. 2014, 206, 68–78. [CrossRef] [PubMed]
2. Martins, F.M.; dos Neto, J.M.R.; da Cunha, C.J. Mineral phases of weathered and recent electric arc furnace dust. J. Hazard. Mater. 2008, 154, 417–425. [CrossRef]
3. Suetens, T.; Klaasen, B.; Van Acker, K.; Blanpain, B. Comparison of electric arc furnace dust treatment technologies using exergy efficiency. J. Clean. Prod. 2014, 65, 152–167. [CrossRef]
4. Youcai, Z.; Stanforth, R. Extraction of zinc from zinc ferrites by fusion with caustic soda. Miner. Eng. 2000, 13, 1417–1421. [CrossRef]
5. Wang, C.; Guo, Y.F.; Wang, S.; Chen, F.; Tan, Y.J.; Zheng, F.Q.; Yang, L.Z. Characteristics of the reduction behavior of zinc ferrite and ammonia leaching after roasting. Int. J. Miner. Metall. Mater. 2020, 13, 1417–1421. [CrossRef]
6. Lupi, C.; Pilone, D. Effectiveness of saponified D2EHPA in Zn(II) selective extraction from concentrated sulphuric solutions. Miner. Eng. 2020, 150, 106278. [CrossRef]
7. Zhu, K.; Cheng, C.Y. A Study on zinc recovery from leach solutions using ionquest 801 and its mixture with D2EHPA. Miner. Eng. 2012, 39, 117–123. [CrossRef]
8. Zhou, K.; Wu, Y.; Zhang, X.; Peng, C.; Cheng, Y.; Chen, W. Removal of Zn(II) from manganese-zinc chloride waste liquor using ion-exchange with D201 resin. Hydrometallurgy 2019, 190, 105171. [CrossRef]
9. Rudnik, E. Recovery of zinc from zinc ash by leaching in sulphuric acid and electrowinning. Hydrometallurgy 2019, 188, 256–263. [CrossRef]
10. Youcai, Z.; Chenglong, Z.; Youcai, Z.; Chenglong, Z. Electrowinning of Zinc and Lead from Alkaline Solutions. In Pollution Control and Resource Reuse for Alkaline Hydrometallurgy of Amphoteric Metal Hazardous Wastes; Springer: Berlin/Heidelberg, Germany, 2017. [CrossRef]
11. Halli, P.; Hamuyuni, J.; Revitzer, H.; Lundström, M. Selection of leaching media for metal dissolution from electric arc furnace dust. J. Clean. Prod. 2017, 164, 265–276. [CrossRef]
12. Kukurugya, F.; Vindt, T.; Haviš, T. Behavior of Zinc, Iron and Calcium from Electric Arc Furnace (EAF) dust in hydrometallurgical processing in sulfuric acid solutions: Thermodynamic and kinetic aspects. Hydrometallurgy 2015, 154, 20–32. [CrossRef]
13. Montenegro, V.; Agatzini-Leonardou, S.; Oustadakis, P.; Tsakiridis, P. Hydrometallurgical treatment of EAF dust by direct sulphuric acid leaching at atmospheric pressure. Waste Biomass Valoriz. 2016, 7, 1531–1548. [CrossRef]
14. Rudnik, E. Investigation of industrial waste materials for hydrometallurgical recovery of Zinc. Miner. Eng. 2019, 139. [CrossRef]
15. Teo, Y.Y.; Lee, H.S.; Low, Y.C.; Choong, S.W.; Low, K.O. Hydrometallurgical extraction of Zinc and Iron from Electric Arc Furnace Dust (EAFD) Using Hydrochloric Acid. J. Phys. Sci. 2018, 29, 49–54. [CrossRef]
16. Kusumaningrum, R.; Fitroturokhmah, A.; Sinaga, G.S.T.; Wismogroho, A.S.; Widayatno, W.B.; Diguna, L.J.; Amal, M.I. Study: leaching of zinc dust from electric arc furnace waste using oxalic acid. In IOP Conference Series: Materials Science and Engineering; IOP Publishing: Bristol, UK, 2019. [CrossRef]
17. Halli, P.; Hamuyuni, J.; Leikola, M.; Lundström, M. Developing a sustainable solution for recycling electric arc furnace dust via organic acid leaching. Miner. Eng. 2018, 124, 1–9. [CrossRef]
18. Dutra, A.J.B.; Paiva, P.R.P.; Tavares, L.M. Alkaline leaching of zinc from electric arc furnace steel dust. Miner. Eng. 2006, 19, 478–485. [CrossRef]
19. Palimaka, P.; Pietrzyk, S.; Stepień, M.; Cieckło, K.; Nejman, I. Zinc Recovery from steelmaking dust by hydrometallurgical methods. Metals (Basel) 2018, 8, 547. [CrossRef]
20. Zhang, D.; Ling, H.; Yang, T.; Liu, W.; Chen, L. Selective Leaching of Zinc from Electric Arc Furnace dust by a hydrothermal reduction method in a sodium hydroxide system. J. Clean. Prod. 2019, 224, 536–544. [CrossRef]
21. Al-Makhadmeh, L.A.; Batitha, M.A.; Al-Harahsheh, M.S.; Altarawneh, I.S.; Rawawdeh, S.E. The effectiveness of Zn leaching from EAFD using caustic soda. Water. Air. Soil Pollut. 2018, 229, 33. [CrossRef]
22. Ma, A.; Zheng, X.; Shi, S.; He, H.; Rao, Y.; Luo, G.; Lu, F. Study on recovery of Zinc from metallurgical solid waste residue by ammoniacal leaching. In Minerals, Metals and Materials Series; Springer: Berlin/Heidelberg Germany, 2019. [CrossRef]
23. Zhang, D.; Zhang, X.; Yang, T.; Rao, S.; Hu, W.; Liu, W.; Chen, L. Selective leaching of Zinc from blast furnace dust with mono-ligand and mixed-ligand complex leaching systems. Hydrometallurgy 2017, 169, 219–228. [CrossRef]
24. Yang, T.; Rao, S.; Zhang, D.; Wen, J.; Liu, W.; Chen, L.; Zhang, X. Leaching of low grade zinc oxide ores in nitrilotriacetic acid solutions. Hydrometallurgy 2016, 161, 107–111. [CrossRef]
25. Herrero, D.; Arias, P.L.; Güemez, B.; Barrio, V.L.; Cambra, J.F.; Requies, J. Hydrometallurgical process development for the production of a zinc sulphate liquor suitable for electrowinning. *Miner. Eng.* 2010, 23, 511–517. [CrossRef]

26. Eksteen, J.J.; Oraby, E.A.; Tanda, B.C. A conceptual process for copper extraction from chalcopyrite in alkaline glycinate solutions. *Miner. Eng.* 2017, 108, 53–66. [CrossRef]

27. Oraby, E.A.; Eksteen, J.J.; Karrech, A.; Attar, M. Gold extraction from paleochannel ores using an aerated alkaline glycine lixiviant for consideration in heap and in-situ leaching applications. *Miner. Eng.* 2019, 138, 112–118. [CrossRef]

28. Prasetyo, E. Monosodium Glutamate For Simple Photometric Iron Analysis. In *IOP Conference Series: Materials Science and Engineering*; IOP Publishing: Bristol, UK, 2018. [CrossRef]

29. Prasetyo, E. Simple method of copper analysis using monosodium glutamate and its application in ore analysis. *Mineralogia* 2012, 43, 137–146. [CrossRef]

30. Prasetyo, E.; Bahfie, F.; Al Muttaqii, M.; Handoko, A.S.; Nurjaman, F. Zinc extraction from electric arc furnace dust using amino acid leaching. In *AIP Conference Proceeding*; AIP Publishing: College Park, MA, USA, 2020.

31. Gustafsson, J.P. Visual MINTEQ Ver. 3.1. Available online: https://vminteq.lwr.kth.se/visual-minteq-ver-3-1/ (accessed on 4 August 2019).

32. Leonard, M. *Vogel’s textbook of quantitative chemical analysis.* *Endeavour* 1990, 14, 100. [CrossRef]

33. Anderegg, G. Critical survey of stability constants of EDTA complexes. In *Critical Survey of Stability Constants of EDTA Complexes*; Pergamon: Oxford, UK, 1977. [CrossRef]

34. Anderegg, G. Critical survey of stability constants of Nta complexes. *Pure Appl. Chem.* 2013, 54, 2693–2758. [CrossRef]

35. Kiss, T.; Sovago, L.; Gergely, A. Critical survey of stability constants of complexes of glycine. *Pure Appl. Chem.* 2007, 63, 597–638. [CrossRef]

36. Dickinson, C.F.; Heal, G.R. Solid-liquid diffusion controlled rate equations. *Thermochim. Acta* 1999, 340, 89–103. [CrossRef]

37. Rao, S.; Yang, T.; Zhang, D.; Liu, W.F.; Chen, L.; Hao, Z.; Xiao, Q.; Wen, J.F. Leaching of low grade zinc oxide ores in nh4cl-nh3 solutions with nitrilotriacetic acid as complexing agents. *Hydrometallurgy* 2015, 158, 101–106. [CrossRef]

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