

Article



# Zinc Recovery from Steelmaking Dust by Hydrometallurgical Methods

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**Abstract:** Hydrometallurgical recovery of zinc from electric arc furnace dust was investigated on a laboratory scale, using aqueous sodium hydroxide solution as a leaching agent. Special attention was paid to the effect of NaOH concentration, temperature and liquid/solid phase ratio on the zinc leachability. It was found that all tested factors increased the leachability, with the maximum efficiency of 88% obtained in a 6 M NaOH solution at a temperature of 80 °C and the liquid/solid phase ratio of 40. The test results confirmed the high selectivity of the zinc leaching agent. In spite of this, complete recovery of zinc from steelmaking dust has proved to be very difficult due to the occurrence of this element in the form of stable and sparingly soluble  $ZnFe_2O_4$  ferrite. Purification of the solution by cementation and electrolysis gave zinc of purity 99.88% in powder form.

Keywords: steelmaking dust; zinc recycling; alkaline leaching; electric arc furnace

## 1. Introduction

One of the commonly used techniques to protect steel surfaces from corrosion is galvanizing. Almost 50% of the world's zinc production is consumed for this purpose. Worn steel elements are usually sent to metallurgical plants and are subjected to a remelting process in electric arc furnaces (EAF). During this process, zinc is evaporated, oxidized and then, as solid ZnO, transferred to dedusting devices. One ton of smelted steel scrap produces about 15–25 kg of dust, in which the content of zinc is 15–40% [1]. The dusts additionally contain a significant amount of iron (about 20%) and other elements such as cadmium, nickel, chromium, manganese, carbon, tin, antimony and copper. The storage of this type of waste is associated with serious threats to the natural environment, as metals can leach out to surface and groundwater, contaminating the environment [2]. Moreover, following accumulation in organisms such as plants and animals, they reach humans. Since the metals do not decompose, they can bioaccumulate once they are absorbed [3]. Therefore, the waste should be recycled to the steelmaking process, which is not currently possible due to the high content of zinc. On the other hand, a significant iron content eliminates the possibility of treating the steel dust as a raw material for the traditional hydrometallurgical zinc manufacturing process.

Current technologies for steel dust processing are focused on pyrometallurgical methods, and consist mainly of high-temperature reduction of ZnO contained in dust and secondary oxidation of Zn in the gas phase. Figure 1 shows the pyro-and hydrometallurgical methods for steelmaking dust processing currently used on an industrial and pilot scale. The Waelz kiln process remains the predominant method for processing dust, with over 85% of the market [4]. Any other pyrometallurgical process, indicated in Figure 1, results in zinc in the form of ZnO, which is fed to the traditional process of sulfate electrolysis or Imperial Smelting Process (ISP).



Figure 1. Schematic presentation of electric arc furnace dust treatment [3–7].

The attempt to implement a comprehensive and waste-free process described in References [3,5] was undertaken by a plant in Poland. The pilot installation that has been launched allows two fully usable products to be made in a shaft furnace. The first product is Zn, in the form of ZnO, removed from properly formed briquettes and captured in dedusting devices. The second product includes zinc-free sintered briquettes, which can be used in steelmaking. In this way, no waste is generated in the process.

Hydrometallurgical processes are not as popular as pyrometallurgical methods, but there are several installations converting EAF dust on an industrial scale. One of the currently used technologies is the EZINEX process, which involves ammonia leaching of dust, cementation-based purification of the solution and zinc separation through electrolysis. In recent years, the process has been modified with high-temperature pre-treatment of zinc-bearing material (fuming in an induction furnace). As a result of this process, raw zinc oxide is subjected to a hydrometallurgical treatment, and the integrated technology is called ENDUTEC/EZINEX [6]. Another process (Modified Zincex Process) is a modification and simplification of the former process (Zincex) using atmospheric leaching  $(H_2SO_4 \text{ solution})$ , solvent extraction and conventional electrowinning, in order to resolve impurity difficulties and recover Special High Grade (SHG) zinc ingots [8]. In turn, Recupac has developed a patented recycling process to extract marketable iron and zinc compounds from EAF dust using proven hydrometallurgical technology. The zinc and iron are recovered, and the iron oxide is used to make industrial pigments. Many processes developed on a laboratory, pilot or major scale have not been implemented in industry, or, for indefinite reasons, have been delayed after several years (Cebedeau, Cardif, AMAX) [9]. The biggest problem in hydrometallurgical processes is the maximization of zinc recovery with simultaneous inhibition of iron extraction into the solution [10]. Since zinc occurs in the dust as ZnO and ZnFe<sub>2</sub>O<sub>4</sub>, the key problem is how to extract it from the zinc ferrite and transfer it to

the solution. If zinc in the dust is mainly present in the form of ZnO, high leaching efficiency is to be expected. If, in turn, zinc is in large part bound in  $ZnFe_2O_4$ -a phase of very high stability-then zinc recovery may be low [10]. Generally, hydrometallurgical processes involve leaching of steelmaking dust and extraction of zinc from the solution in a metallic form or as zinc compounds. The main goals are to obtain the end product of the highest purity, develop a method for the management of intermediate products formed during hydrometallurgical processing, and reduce zinc content in the processed EAF dust to a level that would allow its recycling for use in the steelmaking industry.

Until now, most of the attention in studies of the leaching process has been paid to the use of sulfuric acid and caustic soda as leaching agents. The advantages of using sulfuric acid include high dissolution kinetics, availability and a modest price, as well as the possibility of using low concentrations and a well-known process of electrolysis to obtain metallic zinc, with regeneration of the acid. Conversely, the biggest disadvantage of using sulfuric acid as a leaching agent is the lack of selectivity towards other metals found in steelmaking dust. This applies mainly to iron, which can sometimes have a content in steelmaking dust much higher than that of zinc. This results in high consumption of the leaching agent combined with the need to purify the leach solution from significant amounts of Fe, which is both complex and difficult. Iron hinders the cementation of copper, cadmium and cobalt [9]. It is also necessary to control pH to avoid hydrolysis and Zn(OH)<sub>2</sub> precipitation.

The main advantage of using an alkaline leaching agent is its selectivity towards zinc over iron, which practically eliminates the difficult and complicated process of iron removal from the electrolyte. It is also possible to separate lead from zinc, which may be used as an additional intermediate product and raw material for the recovery of this metal. Therefore, the use of NaOH as a leaching agent seems to be a promising alternative for the processing of steelmaking dust, although alkaline leaching agents also have some disadvantages. The most important include the high concentration necessary to obtain significant zinc leaching efficiency, a higher price compared to sulfuric acid, and the difficult recovery of the solution.

Despite many problems, the hydrometallurgical methods of steelmaking dust processing are much more attractive due to lower operating costs and energy expenditure. Problems related to the high zinc leaching rate and effective extraction from the solution are still a challenge.

Results of research on EAF dust leaching in NaOH solutions that have been published in recent years indicate a strong dependence of some parameters on leaching efficiency. Mordoğan et al. [10] investigated leaching in solutions with a low concentration of NaOH (3.75 M) at ambient temperatures. The effects were poor; only 21.45% of zinc was extracted to the solution. Xia and Pickles [11] and Dutra et al. [12] obtained leaching efficiency of about 74% at higher concentrations of NaOH (6 M) and temperature (90 °C). Orhan [13] attempted to use even more concentrated solution and a higher temperature (10 M NaOH, 95 °C), and achieved up to 85% Zn recovery. Previous results show that the sodium hydroxide concentration and leaching temperature have significant influence on the dissolution of zinc from the EAF dust. Other parameters such as liquid to solid ratio, particle size and agitation speed were also investigated in the above research, but it was found that these parameters had relatively insignificant influence on the extraction process.

Therefore, this study was undertaken to determine the optimal conditions for zinc extraction from EAF dust using caustic soda solutions. The results of the cementation process and electrolysis of the chosen solution, after leaching of dust, were also obtained.

#### 2. Methods

#### 2.1. Leaching

The leaching process was carried out in two stages. The aim of the first stage, involving leaching in water, was to remove chlorine from the material. This is an indispensable step because, during the main leaching process, the presence of chlorine (and also of Na, K, Ca) in the dust causes high consumption of the leaching agent and risk of harmful chlorine being released during electrolysis. The liquid/solid

ratio during leaching was 20 and the temperature was kept at 60 °C. The process lasted 120 min using mechanical stirring at 400 rpm. After water leaching, the material was vacuum filtered. Then, after washing and drying, the residue was subjected to alkaline leaching. The aim of this operation was to transfer as much zinc as possible from the EAF dust to the solution. Sodium hydroxide was used as the leaching agent. The variable parameters investigated for the leaching process included NaOH concentration, temperature and the liquid/solid ratio (L/S). These parameters are summarized in Table 1.

Table 1. The parameter values for alkaline leaching of Electric Arc Furnace (EAF) dust.

Parameter	Value			
NaOH Concentration, M	2, 4, 6			
Temperature, °C	20, 50, 80			
Liquid–solid ratio, cm <sup>3</sup> /g	10, 20, 30, 40			
Stirring speed, rpm	400			
Time, min	120			

The experiments were carried out in an 800 mL glass beaker placed on a heating plate. The temperature was controlled to the nearest 1 °C and kept constant using a sensor, protected by a quartz shield, placed in the solution. The duration of each leaching experiment was 120 min. During this process, 4 mL samples of the solution were taken at 5, 15, 30, 60 and 120 min. Each sample taken was filtered and then subjected to chemical analysis using X-ray fluorescence (XRF) (MiniPal 4, PANalytical B.V., Espoo, Finland). Analysis of the concentration of metals in the solution was based on a multi-point calibration curve. The efficiency of zinc extraction into the solution was calculated from Equation (1):

$$L_e = (m_{Zn,t} / m_{Zn,0}) \times 100\%$$
(1)

where:  $L_e$  (%) is the zinc extraction efficiency,  $m_{Zn,t}$  (g) is the zinc content in the solution after leaching time t, and  $m_{Zn,0}$  (g) is the zinc content in the sample before leaching.

#### 2.2. Cementation

The cementation-based purification of the solution was carried out with a 99.95% pure zinc metal plate, keeping the process temperature at a level of 60 °C and stirring the solution with a magnetic stirrer. The cementation time was 70 min. During the experiment, samples for analysis were taken every 3 min (during the first 15 min of the experiment), then every 5 min (up to 30 min of the experiment) and at 50 and 70 min.

#### 2.3. Electrolysis

The electrolysis was carried out for 90 min at a current density of  $250 \text{ A/m}^2$  and a temperature of 25 °C. Graphite was used as the material of the anode, while the cathode was made of stainless steel. The current efficiency was calculated from Equation (2):

$$\eta = (m_{Zn}/(k \times I \times t)) \times 100\%$$
<sup>(2)</sup>

where:  $\eta$  (%) is the current efficiency,  $m_{Zn}$  (g) is the mass of zinc after electrolysis, k is the electrochemical equivalent, I (A) is the current and t (sec) is the electrowinning time.

#### 3. Materials

The steelmaking dust used in the study was obtained from a Polish steel plant. All the experiments were carried out on dust in the form of finely divided fractions (less than 100  $\mu$ m) with a moisture content of 5% (estimated from a 24 h drying test at 105 °C). The chemical and mineralogical compositions of the dust were determined by Microwave Plasma Atomic Emission Spectroscopy

(MP-AES) (MP-AES 4200, Agilent, Penang, Malaysia), and Scanning Electron Microscopy (SEM) (Hitachi, Tokyo, Japan) and X-ray diffraction (XRD) (MiniFlex II, Rigaku, Tokyo, Japan), respectively. The composition and morphology of the dust strongly depends on the quality of the scrap and on the mode of operation of the electric furnace [3]. Typical EAF dust particles obtained by scanning electron microscopy with analysis of a selected particle, are presented in Figure 2.



**Figure 2.** Spherical Electric Arc Furnace (EAF) dust particles and chemical analysis using Scanning Electron Microscope (SEM).

The presence of spherical particles rich in iron, oxygen and zinc was observed. The results of the chemical composition of EAF dust are given in Table 2. The X-ray diffraction pattern of the EAF dust is shown in Figure 3.



Table 2. Chemical composition of the main elements present in the dust.

Figure 3. X-ray diffraction analysis of EAF dusts.

Five major phases were identified: franklinite, magnetite, zincite, and sodium and potassium chlorides. The identified chlorides were a residue generated from chlorinated rubber coatings present in plastic admixtures (including polyvinyl chloride and synthetic rubber) in the charge melted to generate the EAF dust. These chlorides are volatile under the process conditions used and evaporate easily, which explains their presence in the dust. Many other compounds have also been identified in the dust [11,12,14,15], but being present in insufficient quantities, these were difficult to detect.

### 4. Results

#### 4.1. Leaching

In this study, the effect of 2, 4 and 6 M NaOH on the zinc extraction efficiency was investigated. The tests were limited to a concentration of 6 M, because the viscosity of solutions with higher NaOH content increased too much, hindering further processing operations.

The main reactions occurring during leaching of dust can be written in the form of reactions (3) and (4):

$$ZnO_{(s)} + 2NaOH_{(aq)} \rightarrow Na_2ZnO_{2(aq)} + H_2O_{(l)}$$
(3)

$$PbO_{(s)} + 2NaOH_{(ag)} \rightarrow Na_2PbO_{2(ag)} + H_2O_{(l)}$$

$$\tag{4}$$

The high stability of ferrites means that only some of them can dissolve in NaOH solutions, according to reaction (5) [10]:

$$ZnFe_2O_4 + 2NaOH_{(aq)} \rightarrow Na_2ZnO_{2(aq)} + H_2O_{(l)} + Fe_2O_{3(s)}$$
(5)

Sample leaching curves for all NaOH concentrations, limit values for temperature and L/S conditions (20 and 80  $^{\circ}$ C and L/S of 10 and 40) are shown in Figure 4.



**Figure 4.** Leaching efficiency for zinc at different leaching conditions of NaOH concentration (2, 4, 6 M), temperature (20 and 80 °C) and liquid/solid (L/S) ratio (10 and 40).

The leaching results obtained for the limit parameter values lead to the conclusion that leaching in 2 M NaOH and L/S = 10 offers a very modest process efficiency, giving a maximum of 24%. During leaching at 20 °C in 4 and 6 M solutions (L/S = 10), a continuous increase in the zinc extraction was observed, which means that these solutions have the potential to dissolve more Zn, but in a time much longer than that used in the experiments. Increasing the temperature to 80 °C increases the

zinc extractions to 40 and 45% for 4 and 6 M NaOH, respectively, after only 30 min. Further increases in the leaching time do not result in higher zinc extractions. For L/S = 40, the extractions are much higher, giving 68 and 88%, after 120 min in 6 M NaOH, at 20 °C and 80 °C, respectively. Table A1 in Appendix A summarizes all the results of zinc extraction and its concentration obtained after 120 min for each leaching test. Figures 5 and 6 show these results in graphical form.



**Figure 5.** Leaching efficiency for zinc as a function of L/S ratio for different NaOH concentrations and temperature.



**Figure 6.** Zinc concentration as a function of L/S ratio for different NaOH concentrations and temperature.

The results obtained show that zinc leaching efficiency depends on all the examined factors, i.e., time, temperature, NaOH concentration and the solid phase content in the leaching solution. At each temperature, the increase in concentration from 2 M to 4 M NaOH increased zinc extraction by a dozen percentage points (13.3–23.4). For further increase in NaOH concentration, the efficiency gains were much lower (0.9–7.9 pp). This may be due to either higher viscosity of the solution

and the resultant difficulty in transporting the reactants to and from the reaction surface, or from achieving the maximum extraction under the conditions employed. Additionally, the smallest effect of temperature was observed in solutions with a high solid phase content (L/S = 10). In this case, raising the temperature by 30 °C increased the efficiency of the zinc leaching process by a maximum of 6 pp. In each subsequent leaching test, in which the liquid phase content was increased with respect to the solid phase, the effect of temperature was stronger, and the maximum increase in leaching efficiency was about 15.6 pp for L/S = 30 and 2 M NaOH. Although increasing the liquid phase content in the leached slurry had a beneficial effect on the dust leaching efficiency in terms of metal extraction into the solution in general, the concentration of zinc in the solution actually decreased with increased L/S. The increase of L/S from 10 to 40 in the case of leaching in 2 M NaOH at 80 °C resulted in the drop of zinc concentration from 7.93 g/dm<sup>3</sup> to 5.78 g/dm<sup>3</sup>, i.e., about 27%. At the highest NaOH concentration of 6 M, the decrease in zinc concentration associated with dilution of the solution was 52% (from 15.19 to 7.26 g/dm<sup>3</sup>). Erdem et al. [16] observed that waste materials containing zinc in the form of ZnFe<sub>2</sub>O<sub>4</sub> were difficult to leach because franklinite is very stable and poorly soluble in most types of alkaline solutions. They achieved a maximum leachability of 21.47% during leaching at 25 °C for 120 min, using a stirring speed of 300 rpm, L/S = 5 and NaOH concentration of 15%. In the present study, it was found that high solid phase content results in low leaching efficiency of Zn, therefore the results obtained by Erdem et al. [16] appear to be lower than those which can be achieved. The maximum zinc extraction (88.0%) in this study was obtained for 6 M NaOH at 80 °C and L/S = 40. Figure 7 presents a diffractogram of EAF dust samples before and after leaching under selected process conditions.



**Figure 7.** X-ray diffraction analysis of EAF dusts before and after leaching (in 4 and 6 M NaOH, L/S = 40, temp. = 80 °C, time = 120 min).

XRD analysis of the residue obtained after leaching in 4 M NaOH at 80 °C with L/S = 40 showed that virtually all zinc, present in the form of ZnO, was removed from the dust. An increase in NaOH concentration from 4 to 6 M increased the leaching efficiency by a very small extent only, i.e., by 3 pp. It was, however, an indication that zinc in the form of  $ZnFe_2O_4$  was being dissolved according to

reaction (5), although this process was slow. The dissolution of franklinite was suggested by the lower intensity of franklinite peaks in the diffractogram for the sample leached in 6 M NaOH at 80 °C with L/S = 40 (Figure 7). Analysis of the concentrations of other elements in the leach solution confirmed good selectivity of NaOH towards zinc. In none of the conducted leaching tests did the concentration of iron in the solution exceed 20 mg/dm<sup>3</sup>. The only dust component that could dissolve in larger amounts according to reaction (4) is lead. Its concentration under the optimal for zinc leaching conditions (6 M NaOH at 80 °C with L/S = 40) did not exceed 550 mg/dm<sup>3</sup>, though even this concentration is quite significant and might affect the process of electrolysis (i.e., contaminate the zinc cathode). The concentration of other metals (Cr, Mn, Cd) in all leaching tests was low and generally below 40 mg/dm<sup>3</sup>, or much lower. The tested leaching parameters had no specific effect on the concentrations of these metals. Table 3 outlines the results of the solution analysis after leaching at 80 °C (6 M NaOH; L/S = 10). This particular solution was selected for further treatment, as it had the highest concentration of zinc.

Table 3. Leach solution composition after 120 min (6 M NaOH; 80  $^{\circ}$ C; L/S = 10).

Element	Zn	Cr	Mn	Fe	Cd	Pb
mg/dm <sup>3</sup>	15190	40	12	20	12	420

## 4.2. Cementation

The removal of undesirable elements from the electrolyte is necessary for the final stage of processing: Electrolysis. In the electrolysis process, any metal more noble than zinc will be deposited on the cathode, contaminating the zinc cathode and reducing the current efficiency. Therefore, it is important that the solution used for electrolysis contains the least amount of undesirable metals. The cementation process can be written in the general form of Equation (6):

$$n_2 \mathrm{Me}^{n_1 +} + n_1 \mathrm{M}^0 = n_2 \mathrm{Me}^0 + n_1 \mathrm{M}^{n_2 +}$$
(6)

where:  $n_1$ ,  $n_2$  is the ion charge; Me is the cementated metal; and M is the cementing metal.

According to the above reaction it is possible to remove more noble metals with zinc, and in this particular case, mainly a significant amount of lead. Figure 8 shows the changes in the concentration of zinc and lead throughout the 70-min cementation process.



Figure 8. Concentration of Pb (blue) and Zn (red) during the cementation process.

As can be seen from Figure 8, practically all lead was removed in just 50 min. The concentration of other metals remained unchanged during the cementation process. The lead concentration in the solution after cementation (<10 mg/dm<sup>3</sup>) was found to be low enough to not affect zinc electrolysis. It should be noted here for comparison that Orhan [13] studied the effectiveness of the cementation process using zinc powder. Despite the selection of optimal process parameters, even after 3 h of cementation, lead was removed from the solution to a level of 100–120 mg/dm<sup>3</sup>. Furthermore, Mordogan et al. in Reference [10], who also used zinc powder, removed lead from the solution to the level of about 210 mg/dm<sup>3</sup> using zinc powder.

The composition of the post-cementation solution which was used for electrolysis is presented in Table 4.

Table 4. The composition of the solution after the cementation process.

Element	Zn	Cr	Mn	Fe	Cd	Pb
mg/dm <sup>3</sup>	16200	40	12	20	12	<10

#### 4.3. Electrolysis

The reactions which occur during the process of electrolysis can be written in the following form:

$$A(+): 2OH^- \to 1/2O_2 + H_2O + 2e$$
 (7)

$$K(-): Zn(OH)_4^{2-} + 2e \rightarrow Zn + 2OH^- + H_2O$$
 (8)

In total, the process of metallic zinc deposition from the solution can be written as:

$$Na_2ZnO_2 + H_2O \rightarrow Zn + 2NaOH + 1/2O_2$$
(9)

Since the concentration of elements in the electrolyte was low (Table 4), the process of electrolysis was straightforward. The result was a spongy powder, having a morphology as shown in Figure 9.



Figure 9. Images of zinc powder deposited on cathode; SEM.

The process efficiency with the applied current density of  $250 \text{ A/m}^2$  was 86%. A similar value under similar experimental conditions was obtained by Zhang et al. [17], but there was no mention of the structure of the cathode deposit and purity of the obtained zinc. The chemical composition of the zinc deposited on the cathode from the present study is given in Table 5.

Table 5. Chemical composition of the cathodic zinc.

Compound	Concentration, %				
Fe	0.064				
Cd	0.051				
Zn	99.88				

The deposit obtained in this study was easy to remove from the cathode surface and, as shown by the analysis in Table 5, had a high degree of purity. The deposit was repeatedly rinsed with distilled water before the analysis was free of sodium.

#### 5. Discussion

Currently, in terms of development, the hydrometallurgical technologies are lagging behind the pyrometallurgical processes used in most industrial installations for the processing of steelmaking dust. Nevertheless, studies of hydrometallurgical techniques for zinc recycling from this type of waste are still ongoing and have become important for several reasons:

- Hydrometallurgical processes are environmentally friendly, generating practically no air pollution.
- Installations are much smaller in size compared to pyrometallurgical plants, and as such require lower investment and operating costs.
- Hydrometallurgical processes are applicable in the recovery of other valuable metals.
- The product obtained from leaching is an iron-bearing material that can be recycled and reused in the steelmaking process (if the zinc content will be at the minimum level).

As shown by this study, leaching with NaOH solution is selective towards zinc, and the process proceeds with a high yield of up to 88.0%. However, the presence of zinc in the form of  $ZnFe_2O_4$  ferrite poses some problems in regards to its complete recovery from steelmaking dust. Studies have shown that even high concentrations of NaOH (6 M), high temperature (80 °C), and adequate excess of leaching solution in relation to the solid phase (L/S = 40) does not remove all of the zinc present. Although the results of the XRD analysis of the leach residue, generated in 4 and 6 M NaOH at a temperature of 80 °C, indicated that zinc ferrite dissolution does occur, the rate of this reaction is low. Xia and Pickles [11] found that even in a highly aggressive leaching environment (10 M NaOH, temperature of 93 °C, leaching time of 3 h), the extraction yield of zinc from the zinc ferrite is only 9%. A solution to this problem might be sintering of dusts with Na<sub>2</sub>CO<sub>3</sub> before the leaching process, in order to decompose the zinc ferrite according to reaction (10), as suggested by Holloway et al. [18].

$$ZnFe_2O_4 + Na_2CO_3 \rightarrow ZnO + 2NaFeO_2 + CO_2$$
(10)

Another solution may be to grind the EAF dust with iron powder (mechanochemical reduction). This also enables the breakdown of  $ZnFe_2O_4$ , as demonstrated by Zhang et al. [19]. Such approaches generate additional costs due to the expansion of the installations to include aggregates for material processing, such as mills, furnaces and dryers, but the end product—i.e., an iron-bearing material—is free from zinc (or has only a very low content of this metal), and thus can be returned to the steelmaking process. The results of cementation and electrolysis presented in this study showed that both processes run smoothly to give a high purity final product.

## 6. Conclusions

- 1. Leaching of EAF dust using NaOH solutions confirmed high zinc selectivity.
- 2. All parameters tested (time, temperature, L/S ratio and NaOH concentration) had a positive effect on the efficiency of zinc leaching.
- 3. Increasing the leachate content in relation to the solid phase content increases the leaching efficiency, but also decreases the zinc concentration in the solution. This means that it is necessary to handle larger volumes of solution to obtain the required EAF dust throughput.
- 4. The maximum zinc leaching efficiency of 88.0% was achieved in a 6 M NaOH solution at a leaching temperature of 80 °C and L/S = 40. Complete leaching of zinc from steelmaking dust under the tested conditions was not possible due to the presence of this element in the form of  $ZnFe_2O_4$ .

- 5. The cementation process using metallic zinc enables the removal of lead from the solution to a very low level of concentration (<10 mg/dm<sup>3</sup>).
- 6. The cathode zinc was obtained as spongy-powder and is characterized by high purity (99.88%).

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Conflicts of Interest: The authors declare no conflict of interest.

#### Appendix A

L/S, cm <sup>3</sup> /g	Temp., °C	C <sub>NaOH</sub> , M	Le, %	$C_{Zn}$ , g/cm <sup>3</sup>	C <sub>NaOH</sub> , M	Le, %	$C_{Zn}$ , g/cm <sup>3</sup>	C <sub>NaOH</sub> , M	Le, %	$C_{Zn}$ , g/cm <sup>3</sup>
10	20		16.5	5.4		31.6	10.4		39.5	13.0
	50		21.2	7.0		34.5	11.4		42.0	13.9
	80		24.0	7.9		40.6	13.4		46.0	15.2
	20		28.3	4.7		45.4	7.9	-	51.4	8.5
20	50		39.3	6.4		58.1	10.2		56.0	9.9
	80	2	44.8	7.4	4	68.2	12.2	6	70.3	11.6
30	20		38.1	4.2		55.2	6.6		59.6	6.6
	50		53.6	5.9		70.1	7.7		71.2	7.8
	80		59.8	6.7		79.1	8.7		80.0	8.8
40	20		50.0	4.1		65.1	5.5		68.1	5.6
	50		62.2	5.1		78.1	6.4		79.1	6.5
	80		70.1	5.8		85.0	7.2		88.0	7.2

Table A1. Results of leaching experiments after 120 min.

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