

Article



Optimization of Metal Recovery from MSWI Fly Ash by Acid Leaching: Findings from Laboratory- and Industrial-Scale Experiments

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Abstract: A major part of Swiss fly ashes (FA) from municipal solid waste incineration (MSWI) are treated with the acid fly ash leaching process (FLUWA) in order to recover heavy metals prior to deposition. The FLUWA process uses scrub water from wet flue gas cleaning to leach heavy metals from FA. The leaching efficiency is strongly dependent on the leaching conditions (e.g., pH, Eh, L/S-ratio). This case study presents the optimization of the FLUWA process at the MSWI plant Linth, Switzerland, through determination of ideal process parameters for optimal metal recovery. By means of laboratory- and industrial-scale experiments, the process was adjusted towards a more efficient leaching of Zn, Pb, Cu, and Cd. This included the use of an oxidizing agent (hydrogen peroxide). Laboratory experiments proved to be a powerful tool for simulating process optimizations at industrial scale. An ideal leaching pH of 3.8 was determined and it was observed that the process stability is significantly influenced by the L/S-ratio applied to the leaching process. In the course of the study, the recovery could be improved to 67% Zn, 66% Pb, 30% Cu, and 91% Cd. It can be concluded that for optimal metal recovery the process has to be individually adjusted to the composition of the processed FA and scrub water of each specific FLUWA process.

Keywords: MSWI fly ash; acid leaching; heavy metal recovery; process optimization

1. Introduction

In Switzerland, approximately 80,000 tons of fly ash (FA) from municipal solid waste incineration (MSWI) arise annually [1]. Due to the incineration of heavy metal bearing objects (e.g., electronics or batteries) contained in the waste, the FA is highly enriched in heavy metals. Swiss FA contain up to 6.6 wt.% Zn, 1.9 wt.% Pb, 0.6 wt.% Cu, and 0.05 wt.% Cd [2], and such elevated heavy metal concentrations have been observed in ashes from other countries [3]. Thus, FA comprises a large potential for recyclable metals where mainly the elements Zn, Pb, Cu, and Cd are of interest. In recent years, a research focus was put on the optimization of the heavy metal separation and recovery from FA. Acid leaching was shown to be the most effective method to mobilize heavy metals from FA [4–11].

Since 1995, the acid fly ash leaching process (FLUWA process) has been established in Switzerland. Therein, heavy metals are extracted from FA using scrub water [12–14], a liquid residue from wet flue gas cleaning. The heavy metals are accumulated in the leachate which can later be used for heavy metal recovery. The aim of the FLUWA process is to produce a leachate that contains high concentrations of the metals to be recovered. The metal recoveries achieved in the FLUWA process depend on the properties of the FA as well as on the leaching conditions. The main process parameters that influence metal mobilization are pH value and the oxidation-reduction potential [15]. The leaching pH is defined by the alkalinity of the FA and the amount of acid accumulated from the scrub water. In order to reach sufficient acidic process conditions for mobilizing the heavy



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Copyright: © 2021 by the authors. Licensee MDPI, Basel, Switzerland. This article is an open access article distributed under the terms and conditions of the Creative Commons Attribution (CC BY) license (https:// creativecommons.org/licenses/by/ 4.0/). metals, the addition of additional acid is often necessary. The pH value defines whether the mobilized metals reprecipitate during extraction, making them unavailable for recovery. In order to create ideal conditions for the extraction of redox-sensitive metals, an oxidant (e.g., hydrogen peroxide) is necessary during the FLUWA process. Oxidizing conditions during metal extraction are a prerequisite to suppress the reductive precipitation of elements such as Cu, Pb, and to a minor extent, Cd [16–18]. Other factors influencing FA leaching are temperature and reaction time. A temperature of ca. 60 °C arises from the mixing of FA and scrub water during the FLUWA process. The reaction time is dependent on the number of extraction reactors used. Three reactors connected in series are prevalent in Switzerland. An extraction time of one hour, cumulated from 20 min per reactor, has commonly been established. Filtration of the ash slurry by vacuum-belt filtration yields a metal-depleted filter cake and a metal-enriched leachate. The filter cake has less impact on the environment and can be deposited together with bottom ash on a landfill type D according to the Swiss Waste Ordinance [19]. The metalliferous leachate is precipitated to a hydroxide sludge. The hydroxide sludge can so far only be processed abroad as hazardous waste. With the foundation of SwissZinc AG, a central treatment facility for hydroxide sludge is pursued. Implementation is expected by the end of 2023 [20]. This plant will apply a system based on the FLUREC process to produce high-purity zinc and as byproducts Pb, Cu and Cd in a recoverable form [12]. The prescription of recovering heavy metals from FA and recovery of metals is prescribed from 2026 onwards in Switzerland [19]. Therefore, the investigation of the FLUWA process is of increasing interest to improve heavy metal separation and to estimate the limiting factors of metal depletion.

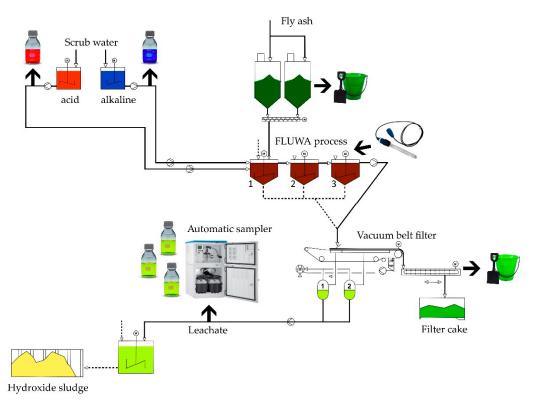
With this study, the optimization of the FLUWA process at the Swiss MSWI plant Linth is shown. The goal of the study was to determine the process parameters for an optimal recovery for Zn, Pb, Cu and Cd. This was achieved by performing experiments both on laboratory and industrial scale. The laboratory-scale experiments allow the preliminary examination of the influence of different parameters on the FA leaching process. As the operation on the industrial process relies on many more variables which cannot all be varied or excluded, the laboratory-scale experiments render it possible to better isolate and examine the influence of selected parameters. These initial experiments also deliver an idea of metal recoveries attainable for the specific FLUWA process. In the course of the study, three industrial-scale experiments were carried out. Such experiments represent the reality of a large-scale operation which runs continuously over hours and days where the input materials may also vary over time. Parallel to each industrial-scale experiment, laboratory-scale experiments were carried out with the same process parameters. This indicates to what extent laboratory-scale experiments can model the industrial FLUWA process. In a first experiment, the "current state" of the FLUWA process was sampled in order to make statements about the operational state and the efficiency of the metal recovery. The successive two industrial-scale experiments were performed to implement and verify the optimized operating parameters determined in the laboratory.

2. Materials and Methods

2.1. FLUWA Process at MSWI Plant Linth: Process Conditions and Composition of FA

A total of 6500 t/y of FA are acid leached in the FLUWA process at MSWI plant Linth. In addition to FA from the plant's own waste incineration, external FA from three other Swiss MSWI plants are treated with the FLUWA process. The FLUWA of the MSWI plant Linth has three reactors operating in series, each with a volume of about 3 m³ with an average residence time of about 45 min (Figure 1).

Approximately 9000 m³ of acid scrub water with a hydrochloric acid load of 1600 t HCl (32%) and 5500 m³ of alkaline scrub water with a sulfur load of 275 t from the wet flue gas cleaning process are available annually for the extraction of the FA. Depending on the experiment, an amount of ca. 2–3 t of FA was leached with 6 m³ of acid and 3 m³ of alkaline scrub water per hour of FLUWA operation. When necessary, HCl 32% was added



for pH control. For the optimization tests, a temporary hydrogen peroxide (H_2O_2 , 30%) feed line was built into reactor 1.

Figure 1. Scheme of the acid fly ash leaching (FLUWA) process and sampling locations during the industrial-scale experiments.

Table 1 summarizes the mean value of the FA composition over the entire project duration and indicates the standard deviation for each element concentration. Compared to the average heavy metal concentrations of Swiss fly ash [2], the FA treated at MSWI plant Linth shows elevated Zn (52,593 mg/kg), Pb (9378 mg/kg), Cu (2671 mg/kg) and Cd (332 mg/kg) concentrations.

Table 1. Mean elemental composition in mg/kg and standard deviation of the fly ash (FA) treated at municipal solid waste incineration (MSWI) plant Linth.

| | Mean Concentration of FA Treated at MSWI Plant Linth | | | |
|----|--|---|--------|--|
| Al | 30,278 | ± | 2268 | |
| Ba | 2152 | ± | 208 | |
| Br | 3255 | ± | 174 | |
| Ca | 144,928 | ± | 8696 | |
| Cd | 332 | ± | 34 | |
| Cl | 97,707 | ± | 2892 | |
| Cr | 684 | ± | 120 | |
| Cu | 2671 | ± | 291 | |
| Fe | 19,274 | ± | 1271 | |
| К | 52,755 | ± | 3590 | |
| Mn | 881 | ± | 67 | |
| Ni | 132 | ± | 16 | |
| Р | 3851 | ± | 117 | |
| Pb | 9378 | ± | 973 | |
| S | 61,619 | ± | 9450 | |
| Sb | 2621 | ± | 273 | |
| Si | 77,689 | ± | 15,562 | |
| Sn | 1447 | ± | 139 | |
| Ti | 9783 | ± | 887 | |
| Zn | 52,593 | ± | 7371 | |

2.2. Chemical Analysis

All solids were dried at 105 °C until constant weight and then ground with a ball mill. Then, 4.0 g of the ground sample was mixed with 0.9 g binder (Fluxana Cereox[®]) and pressed into a powder pill. The analysis of solids was carried out using X-ray fluorescence analysis on a Spectro Xepos ED-XRF. The elements Al, Ca, Cd, Cu, Fe, K, Mn, Pb, S, Sb, Si, Ti and Zn were determined with matrix-adapted calibration for FA or filter cake. The elements Ba, Br, Cl, Cr, Ni, Pb and Sn were determined with the Turboquant method.

The liquid samples (leachates) were diluted with 1% HNO₃ (HNO₃, 1%, VWR Chemicals, AnalaR Normapur) and analyzed with a Spectroblue SOP ICP-OES. The calibration was performed using Merck Multielement Standard IV and the analyses were checked with a check standard (Merck VIII Multielement Standard). The analytical error amounts to $\pm 5\%$ for all elements except Na, K, Ca, Sb and S that showed $\pm 10\%$ error based on multiple measurements of certified standard solutions.

2.3. Laboratory-Scale Experiments

The preliminary laboratory-scale experiments were carried out in order to determine the influence of different conditions and parameters on the heavy metal recovery. It is already known from preliminary studies that mainly pH value, H_2O_2 dosage and L/S-ratio need to be investigated to optimize a specific FLUWA process [15]. Therefore, six different experiments were performed at laboratory scale (Table 2): at extraction pH of 3.8 and 4.5, with a concentration of 0, 20, 40 and 60 L H_2O_2/t ash and with two different L/S ratios of 2 and 3. The experiments were performed twice to ensure reproducibility and the mean value used as result. In addition to the three industrial-scale experiments, laboratory-scale experiments were performed under conditions simulating the specific process conditions as good as possible.

| Parameter | pН | Redox (L/t H ₂ O ₂) | L/S (m ³ /t) |
|--------------|-----|--|-------------------------|
| Experiment 1 | 3.8 | 40 | 2 |
| Experiment 2 | 4.5 | 40 | 2 |
| Experiment 3 | 3.8 | 0 | 2 |
| Experiment 4 | 3.8 | 20 | 2 |
| Experiment 5 | 3.8 | 60 | 2 |
| Experiment 6 | 3.8 | 40 | 3 |

Table 2. Experimental setups for laboratory-scale experiments.

For all laboratory tests, 75 g FA was added to the corresponding amount of acid scrub water required for the given L/S-ratio. For each experiment, 70% acid scrub water and 30% alkaline scrub water (HCl = 1.26-1.42 mol/L) were used. Upon mixing the FA with scrub water, the neutralization heat almost instantly led to a rise in temperature in the reactor to approximately 60–65 °C. A heating stirrer including a thermostat was used to maintain a temperature of 60 $^{\circ}$ C while stirring the mixture at ca. 400 rpm. The neutralization occurring upon mixing also led to a pronounced increase in pH value. The pH was corrected to the desired value (pH 3.8 or 4.5) by the dropwise addition of HCl, 32% (VWR Chemicals, AnalaR Normapur) to the mixture. The dosing of H_2O_2 (30%, Merck, stabilized, for synthesis) was carried out in several small portions within the first 20 min of leaching to simulate the continuous peroxide addition to reactor 1 on the industrial scale. The total leaching time applied was 60 min. The pH and temperature of the suspension were documented as well as the quantities of each reagent added. At the end of the leaching time the mixture was hot filtered. The filter cake was washed twice with deionized water $(50 \text{ mL each}, 18.2 \text{ M}\Omega \cdot \text{cm})$ and the combined leachate was rapidly conserved by diluting it 1:100 (vol/vol) with nitric acid (HNO₃, 1%, VWR Chemicals, AnalaR Normapur).

A main goal of the industrial-scale experiments was to collect time-dependent data on the industrial operation. This in turn allowed to identify and address the problems related to the continuous large-scale industrial process. The experimental setups of the three industrial-scale experiments are shown in Table 3. All material flows were sampled at defined time intervals adapted to the operation (mostly in 30-min intervals). The sampling included FA, scrub water (separated acid and alkaline), filter cake from vacuum belt filter, and leachate (Figure 1). The leachate samples were taken at defined time intervals (sampling in 7 min intervals and mixed to samples of 28 min) by means of an automatic sampler (CSF 48, Endress+Hauser). The sample preparation for the leachates was carried out as soon as possible on site. This includes the uptake of the mass flow (present volume of leachate per unit of time), the filtration and the subsequent stabilization of the sample with HNO₃ (30%, VWR Chemicals, AnalaR Normapur) prior to analysis. In addition, pH value, redox potential, and oxygen content were measured in all three reactors at defined time intervals (usually every 15 min).

Table 3. Experimental setups for industrial-scale experiments.

| Parameter. | pH | Redox (L/t H ₂ O ₂) | L/S (m ³ /t) |
|--------------|-----|--|-------------------------|
| Experiment A | 6 | 0 | 3 |
| Experiment B | 4.5 | 40 | 3 |
| Experiment C | 4 | 40 | 4 |

2.5. Calculation of the Leaching Recovery

When performing laboratory-scale experiments, mass balance is easily determinable because all volumes, concentrations and weights can be determined. Therefore, the calculation of the metal recovery was performed as follows (Formulas (1) and (2)):

$$recovery (\%) = 100 \cdot \left(1 - \frac{c(filter \ cake) [mg \cdot kg^{-1}] \cdot m_{filter \ cake} [kg]}{c(fly \ ash) [mg \cdot kg^{-1}] \cdot m_{fly \ ash} [kg] + c(scrub \ water) [mg \cdot L^{-1}] \cdot V_{scrub \ water} [L]} \right)$$
(1)

where *m* is the mass of filter cake or FA. For the laboratory tests, the recovery can also be calculated via the concentration in the leachate:

$$recovery (\%) = 100 \cdot \frac{c(leachate) \lfloor mg \cdot L^{-1} \rfloor \cdot V_{leachate} \lfloor L \rfloor}{c(fly ash) [mg \cdot kg^{-1}] \cdot m_{fly ash} [kmg] + c(scrub water) [mg \cdot L^{-1}] \cdot V_{scrub water} [L]}$$
(2)

where $V_{leachate}$ is the volume of the leachate at the end of the experiment. The mass balance is then calculated using Formula (3):

$$mass \ balance \ (\%) = 100 \cdot \frac{\left(c(leachate)\left[mg \cdot L^{-1}\right] \cdot V_{leachate}[L]\right) + \left(c(filter \ cake)\left[mg \cdot kg^{-1}\right] \cdot m_{filter \ cake}[kg]\right)}{c(fly \ ash)[mg \cdot kg^{-1}] \cdot m_{fly \ ash}[kg] + c(scrub \ water)[mg \cdot L^{-1}] \cdot V_{scrub \ water}[L]}$$
(3)

As the FLUWA process at MSWI plant Linth is conducted in a continuous process, this weight difference is in most cases not known. The mobilization of the acid-soluble components of the FA (e.g., salts) leads to a reduction of mass of the filter cake. Therefore, the mass loss on the industrial scale is often calculated indirectly by measuring and balancing the "inert elements" (e.g., Al, Si, P, Ca, Ti, Cr, Fe, Ni, Sb and Ba). These elements are hardly mobilized under the respective extraction conditions and therefore accumulate in the filter cake relative to its total mass. Assuming that the absolute amounts of these elements in FA

and filter cake are equal and only their concentrations differ, the average enrichment factor of the inert elements can be used as a proxy for the mass loss (Formulas (4) and (5)).

$$enrichment \ factor = \frac{\sum \left(\frac{c_{inert}(filter \ cake)[mg \cdot kg^{-1}]}{c_{inert}(fly \ ash)[mg \cdot kg^{-1}]}\right)}{number \ of \ inert \ elements \ used}$$
(4)

where *c* is concentration of the respective element.

$$recovery (\%) = 100 \cdot \left(1 - \frac{c(filter \ cake) [mg \cdot kg^{-1}]}{c(fly \ ash) [mg \cdot kg^{-1}] \cdot enrichment \ factor} \right)$$
(5)

3. Results

3.1. Laboratory-Scale Experiments

Figure 2 shows the results obtained in the laboratory-scale experiments. The results of these experiments allowed the identification of optimal parameters for the industrial-scale FLUWA operation at MSWI plant Linth.

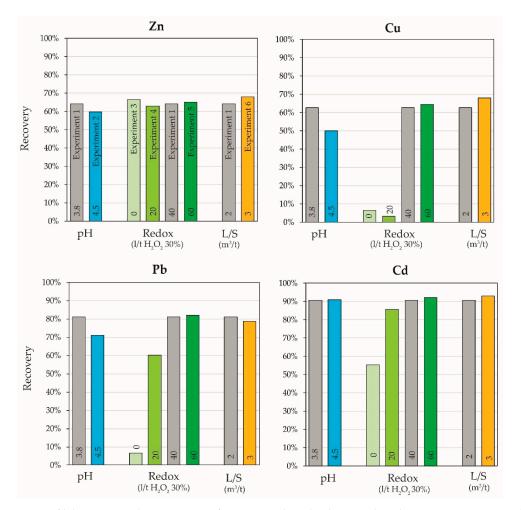


Figure 2. Recoveries of laboratory-scale experiments for Zn, Cu, Pb and Cd to simulate the FLUWA process under different leaching conditions.

The variation of the three parameters pH, redox and L/S ratio throughout the experiments affected the metal recoveries in different ways. A leachate pH value of 4.0 and above at the end of the experiment resulted in diminished recoveries of all four elements studied (Experiment 2). This effect was most pronounced for Cu and Pb. If the pH is adjusted to a value of 3.8 at the end of the leaching process, the recoveries of Cu and Pb are 10% higher (Experiment 1). The oxidizing conditions when adding H_2O_2 to the ash slurry mainly affect the elements Cu and Pb and to a minor extent Cd. Without the addition of an oxidizing agent, 67% Zn and 55% Cd are recovered from FA whereas the recoveries of Pb and Cd are extremely low (<10%) (Experiment 3). The addition of 20 L H_2O_2/t FA increases the recoveries of Pb (60%) and Cd (86%) (Experiment 4). To achieve significant recoveries for Cu, 40 L H_2O_2/t FA are required (63%) (Experiment 1). Table 4 lists the

analytical results of laboratory experiment 1 as well as the mass balance calculated using Formula (3). An addition of more than 40 L H_2O_2 to the ash slurry did not significantly improve Cu recovery (Experiment 5). In addition, the L/S-ratio did not have a large effect on the recoveries measured for the laboratory-scale experiments, at least not in the applied range between L/S 2 and 3 (Experiment 6).

| | Amount of | Elemental concentrations (ppm) | | | |
|--------------------------|----------------------|--------------------------------|---------------------|------------------|----------------|
| | | Cd | Zn | Pb | Cu |
| Fly ash (FA) | 75 g | 364 ± 10 | $47,\!877 \pm 2200$ | $10,080 \pm 275$ | 2942 ± 115 |
| Scrub water | 150 mL | < 0.1 | < 0.1 | < 0.1 | < 0.1 |
| Filter cake [†] | $164\pm1~{ m g}$ | 48 ± 2 | $24,715 \pm 841$ | 2740 ± 23 | 1571 ± 16 |
| Leachate | 180 ± 3 mL $^{++}$ | 128 ± 1 | $11,\!519\pm112$ | 3242 ± 66 | 879 ± 7 |
| Mass balance | | 95% | 93% | 97% | 94% |

Table 4. Mass balance of laboratory-scale experiment 1.

⁺ dry (105 °C). ⁺⁺ comprising 60 mL of deionized water used for rinsing the filter cake.

The optimal leaching conditions determined in the laboratory for the FLUWA process at MSWI plant Linth are as follows: Extraction pH of 3.8, 40 L H_2O_2/t FA and a L/S of 2 (Experiment 1). These parameters served as a basis for the optimization of the industrial FLUWA process.

3.2. Industrial-Scale Experiments

The industrial-scale experiments at the MSWI plant Linth provided time-resolved data over one day of the FLUWA process (Figure 3). The time-resolved sampling of input-FA, leachate and filter cake allowed to determine the metal recoveries (Formulas (4) and (5)). The simultaneous pH monitoring in the last reactor of the FLUWA process helped to identify the potential and limitations of the recovery for the metals Zn, Cu, Pb and Cd.

Experiment A depicted and investigated the "current state" of the FLUWA process before any optimizations. No H_2O_2 was added during this experiment and the pH value and L/S ratio were not adjusted regularly. The concentration of Cu (2648 to 2927 mg/kg) and Cd (250 to 320 mg/kg) in the input-FA were rather constant within the entire day (Figure 3). On the other hand, the concentration of Zn (46,830 to 80,300 mg/kg) and Pb (8743 to 10,680 mg/kg) that enters the FLUWA process with the input-FA was increasing at the beginning of the process and decreasing again with increasing process duration. Depending on the different composition of the FA added to the FLUWA process, the pH value and metal recovery show strong fluctuations over time (Figure 3). The pH value of the ash slurry in reactor 3 started low (pH 2) and gradually increased over time, leveling off at ca. pH 6 at the end of the process. Under these conditions, little Cu and Pb could be recovered throughout the day. Less pH and redox sensitive elements such as Zn and Cu could be recovered better under the prevailing conditions. Especially Zn showed a constant recovery between 58 and 68%, whereas the recovery of Cd fluctuated in the range of 34 to 69%.

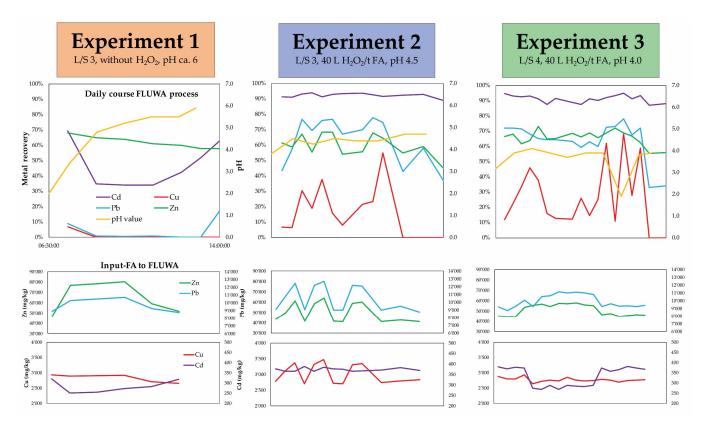


Figure 3. Results of the daily course of the three industrial-scale experiments of the FLUWA process at MSWI plant Linth. The concentrations in the input-FA (mg/kg) and the recoveries (%) of the metals Zn, Pb, Cu and Cd as well as the pH value at the end of the process are shown for each experimental day.

In Experiment B, the initial plan was to be able to better control the pH by a lower FA dosing (increased L/S) and to introduce oxidizing leaching conditions. However, the input-FA dosage could be reduced only slightly due to the large quantities of FA present and the limited capacities in the storage silos. Thus, the L/S-ratio was on a daily average at the same value as in Experiment A (L/S 3). In addition, large concentration fluctuations of Zn (41,080 to 63,860 mg/kg), Pb (9498 to 12,670 mg/kg) and Cu (2699 to 3474 mg/kg) in the input-FA affected the extraction recovery during this experiment. Despite these difficult conditions, it was possible to keep the pH value at a lower level than in Experiment A by the additional dosage of HCl 32%. The pH-value at the end of the leaching process in reactor 3 was maintained at pH 4.5. In addition to the more acidic leaching conditions, oxidizing conditions were established with the addition of $40 \text{ L/t } \text{H}_2\text{O}_2$ to the FLUWA process. The recoveries of Pb, Cu and Cd were significantly better due to the oxidizing, more acidic leaching conditions compared to Experiment A. The daily curves of the leaching recoveries of Zn, Pb and Cu show a clear correlation with the concentrations of the added input-FA. The higher the concentrations of FA, the higher the metal recovery. For Cd, no such correlation is apparent, and the recoveries are very stable at a high level.

Experiment C was again performed under oxidizing conditions using 40 L/t H_2O_2 . However, the input-FA was homogenized compared to experiment B by mixing the FA of the different storage silos before dosing (Zn (44,520 to 57,990 mg/kg), Pb (9222 to 11,100 mg/kg), Cu (2645 to 2929 mg/kg), Cd (269 to 381 mg/kg)). In addition, the FA dosage could be reduced leading to a leaching L/S-ratio of 4. The higher L/S led to the pH value being kept more stable at pH 4.0 during the entire experimental day. Due to the more homogenic FA-input and the more stable process pH-value, the recoveries for Zn and Pb were much more stable during the day. Despite the stable input concentrations, however, there were still exceptionally large fluctuations in the Cu recovery during Experiment C. In addition to the daily course shown in the mean recoveries for Zn, Pb, Cu and Cd over the entire day of the industrial-scale experiments are listed in Table 5. It can be summarized that the recovery of Zn remains at a very stable level (60–70%), irrespective of the adjustments made, whereas for Cd, the recovery could be increased from 52% to >90%. For the more pH- and redox-sensitive elements Pb and Cu, increased recoveries were also achieved with the optimization experiments, but at a lower level (Pb 11% to 66%, Cu 0% to 30%).

Table 5. Mean recoveries of Zn, Pb, Cu and Cd of the industrial-scale experiments of the FLUWA process within one day at MSWI plant Linth.

| Experiment | Zn [%] | Pb [%] | Cu [%] | Cd [%] |
|------------|--------|--------|--------|--------|
| А | 66 | 11 | 0 | 52 |
| В | 60 | 65 | 21 | 92 |
| С | 67 | 66 | 30 | 91 |

The results of the industrial-scale experiments have shown that it is difficult to implement the ideal leaching conditions determined in the laboratory on a large-scale. For this reason, experiments were carried out in the laboratory with the materials (FA and scrub water) and conditions (pH, Eh and L/S) obtained at the large scale-experiments. This allows a direct comparison of laboratory- and industrial-scale experiments (Figure 4). The results show that the recoveries for Cd, Pb and Zn in the laboratory were comparable to those of the industrial-scale experiments. However, despite the detailed reconstruction of the operating conditions, large deviations could be observed in the recovery of Cu. Laboratory-scale experiments show higher leaching recoveries for Cu (7%, 50% and 68%) compared to the corresponding industrial-scale experiments (0%, 22% and 30%).

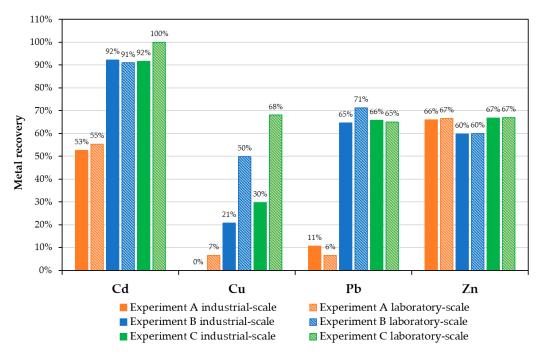


Figure 4. Metal recoveries of the FLUWA process of the three industrial-scale experiments and the corresponding laboratoryscale experiments.

4. Discussion

4.1. Process Stability and Leaching Behaviour of Zn, Cu, Pb, Cd

The recovery of Zn for both laboratory and industrial-scale experiments were between 60% and 70% regardless of the conditions (Figure 4). The use of H_2O_2 had no influence

at all, whereas an increased L/S-ratio and lower pH value slightly but not significantly increased the recovery (Figure 2). Zn recovery thus seems to mainly depend on its input concentration and binding forms. The more Zn is present in FA, the more can usually be mobilized [15].

The redox conditions during extraction are of central importance for the recovery of redox-sensitive elements such as Cu (E°: +0.35 V), Pb (E°: -0.31 V) and Cd (E°: -0.40 V). Without the addition of H₂O₂, the reducing conditions during the FLUWA process led to the cementation reaction of dissolved Cu²⁺, Pb²⁺ and Cd²⁺ on the surface of less noble metals present in the FA, such as Al⁰ (E°: -1.66 V), Zn⁰ (E°: -0.76 V) or Fe⁰ (E°: -0.41 V) (Reaction Equation (6)).

$$3Cu^{2+} + 2Al^0 \rightarrow 3Cu^0 \downarrow + 2Al^{3+} \tag{6}$$

The addition of H_2O_2 to the FLUWA process oxidizes most of the metallic components in FA, resulting in more noble metals such as Cu and Pb being retained in solution and thus recovered (Reaction Equation (7)).

$$2AI^{0} + 3H_{2}O_{2} + 6H^{+} \rightarrow 2AI^{3+} + 6H_{2}O$$
(7)

The recoveries for Cu and Pb are further strongly dependent on the pH value during the leaching process. The time-resolved monitoring of the process conditions in reactor 3 have shown that in a continuous process, the L/S-ratio significantly influences process stability (Figure 3). This is by enhancing or diminishing the effects of naturally occurring fluctuations in both FA and acid scrub water compositions [2]. The FA as well as the scrub water composition depend strongly on the waste input as well as the type and condition of the flue gas cleaning system or dust removal system. They show large variations over time, even in one MSWI plant. In addition to varying heavy metal concentrations, FA also differ in terms of their alkalinity. The carbonate present in the FA thereby buffers the pH of the suspension to >5, which promotes hydroxide precipitation of the metals dissolved from the FA [21]. In order to keep the extracted metals in solution, the pH in the leaching reactors usually has to be lowered with additional HCl (32%). The amount of acid required to reach a certain pH during the FLUWA process varies depending on the FA extracted. The lower the L/S-ratio, the more difficult it is to maintain a constant pH value. With a pH value of \leq 3.8 at the end of the FLUWA process, the hydroxide precipitation of most metals can be kept at a satisfactorily low level. The hydroxide precipitation can be well demonstrated by laboratory-scale experiments (Figure 2). An increase in pH value of 0.7 units between a value of 3.8 and 4.5 led to lower recoveries of Pb and Cu. Cu begins to precipitate as copper hydroxide (or basic copper salts) at a pH of 4.0-4.5 and thus accumulates in the leached filter cake (Reaction Equation (8)) [22].

$$Cu^{2+} + 2OH^{-} \rightarrow Cu(OH)_{2} \downarrow \tag{8}$$

In addition to hydroxide precipitation, Pb can also form poorly soluble sulfates and hence be removed from the leachate by lead-sulfate precipitation [22]. The sulfate supplied by the scrub water does mainly precipitate as gypsum. If the sulfate load in the scrub water exceeds the Ca available from the FA, Pb can form lead sulfate and hence precipitates. This problem can be somewhat alleviated as lead has the ability to form chloride complexes, which in turn are relatively soluble [23]. Alternatively, the scrub water could be treated with a Ca containing reagent in order to reduce the sulfate load before the leaching. As the pH value strongly affects whether the extracted metals re-precipitate and are thus lost, it is important that the pH value of the leachate is closely monitored.

4.2. Comparability Between Laboratory-and Industrial-Scale Experiments

An important aspect of the present project was to verify the transferability of the FLUWA process parameters determined by laboratory-scale experiments to operational practice. Generally, there is a good agreement between the laboratory-scale and the corresponding industrial-scale experiments (Figure 4). Therefore, laboratory-scale experiments

11 of 12

are appropriate to simulate the large-scale FLUWA process. Operating parameters determined in the laboratory for optimal recoveries of heavy metals could further be implemented on a large-scale. In the laboratory, the variation of the L/S-ratio had only a small influence on heavy metal recoveries because there, the complication of temporal change of FA composition present at the industrial scale is omitted. Cu, reacting most sensitive to slight pH value and redox condition changes, is however more challenging to reliably extract on the industrial-scale in comparison to the laboratory-scale leaching. This can be explained by the fact that the pH value in the laboratory reactor can be controlled more precisely than during continuous, large-scale operation.

5. Conclusions

This work has confirmed the importance of a stable process control of the FLUWA in order to achieve a high metal recovery. This is particularly important for the recovery of pH-and redox-sensitive elements such as Cu and Pb. Important parameters for a high metal recovery are the use of a suitable oxidizing agent (e.g., hydrogen peroxide) and precise pH control to assure maintenance of a final pH \leq 3.8 to avoid metal precipitation. Stable process control can be achieved through good mixing of FA as well as elevated L/S-ratio. The optimal leaching parameters have to be determined and adjusted periodically. FLUWA process parameters for one MSWI plant cannot directly be applied to another plant and must be evaluated in laboratory-scale experiments for each specific FLUWA process. With this study it could be successfully demonstrated that this transferability of laboratory-scale experiments into the operational practice is possible.

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