

# Selective recovery of chromium, copper, nickel, and zinc from an acid solution using an environmentally friendly process

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## Abstract

**Purpose** Real electroplating effluents contain multiple metals. An important point related with the feasibility of the bioremediation process is linked with the strategy to recover selectively metals. In this work, a multimetal solution, obtained after microwave acid digestion of the ashes resulted from the incineration of *Saccharomyces cerevisiae* contaminated biomass, was used to recover selectively chromium, copper, nickel, and zinc.

**Results** The acid solution contained 3.8, 0.4, 2.8, and 0.2 g/L of chromium(III), copper, nickel, and zinc, respectively. The strategy developed consisted of recovering copper (97.6%), as a metal, by electrolyzing the solution at a controlled potential. Then, the simultaneous alkalization of the solution (pH 14), addition of H<sub>2</sub>O<sub>2</sub>, and heating of the solution led to a complete oxidation of chromium and nickel recovery (87.9% as a precipitate of nickel hydroxide). After adjusting the pH of the remaining

solution at pH 10, selective recovery of zinc (82.7% as zinc hydroxide) and chromium (95.4% as a solution of cromate) was achieved.

**Conclusion** The approach, used in the present work, allowed a selective and efficient recovery of chromium, copper, nickel, and zinc from an acid solution using a combined electrochemical and chemical process. The strategy proposed can be used for the selective recovery of metals present in an acid digestion solution, which resulted from the incineration of ashes of biomass used in the treatment of heavy metals rich industrial effluents.

**Keywords** Chemical precipitation · Electrolysis · Heavy metals · Recycling · Selective recovery · Chemical speciation

## 1 Introduction

Electroplating effluents often contain excessive concentrations of metals such as chromium, copper, nickel, and zinc that must be removed before being discharged. The environmental threat caused by heavy metals, linked with the lack of efficiency or feasibility of conventional physicochemical technologies, has promoted a big effort research, particularly in the last years, on the development of alternative processes for effective and low cost treatment of metal-bearing industrial effluents (Gadd 2009).

Cells of *Saccharomyces cerevisiae* seem to be a promissory type of biomass for the bioremediation of wastewaters containing heavy metals, as they have the ability of removing a wide variety of metals from synthetic (Chen and Wang 2008; Machado et al. 2008; Machado et al. 2010a; Ruta et al. 2010; Zouboulis et al. 2001) and real effluents (Machado et al. 2010a, b, c; Parvathi and Nagendran 2007; Stoll and Duncan 1997). It was shown

that flocculent brewing strains of *S. cerevisiae* seem particularly appropriated as they have a higher metal uptake than nonflocculent strains (Soares et al. 2002). On the other hand, heat-killed biomass (at 45°C) of brewing yeast cells present higher metal removal than live cells (Machado et al. 2009) and retain their flocculation properties, which overcome the problem of cell separation process after effluent treatment (Machado et al. 2008). In addition, it was demonstrated that flocculent brewing yeast cells are able to treat efficiently complex industrial effluents loaded with several heavy metals (Machado et al. 2010a, b, c).

After heavy metals have been removed by yeast cells from industrial effluents, heavy metal recovery from biomass allows metal recycling since the metals can be reintroduced in the industrial process or resailed. The efficiency of metal recovery is largely dependent on the concentration of metals present in the solution after being extracted from the biomass. Desorption of heavy metals from yeast cells has been attempted by several authors using different eluants such as mineral and organic acids or complexing agents (Ferraz et al. 2004; Strandberg et al. 1981; Wilhelmi and Duncan 1996); however, the lower capacity to concentrate metals strongly limits the feasibility of the recovery process. Alternatively, the amount of yeast biomass loaded with heavy metals (coming from the bioremediation process) can be strongly reduced by incineration; then, the acid digestion of the ashes generates a concentrated metal and acid solution (Machado et al. 2010d).

Usually, industrial effluents contain multiple heavy metals. An important point related to the metals recovery is linked with the strategy to remove selectively metals from the solution. The simultaneous recovery of several heavy metals is indubitably a more ambitious and difficult challenge. Thus, it is not surprising that a limited work has been devoted to this objective. The selective recovery of copper, nickel, and zinc by sequential precipitation using sulfides was described (Tokuda et al. 2008). However, sulfide toxicity and consequently the safety aspects related with its handling strongly impair its practical application (Huisman et al. 2006).

Recently, our group developed a new strategy for recycling selectively heavy metals from yeast cells contaminated with copper, nickel, and zinc, coming from the treatment of real electroplating effluents (Machado et al. 2010d). However, real electroplating effluents can be even more complex and present simultaneously chromium, copper, nickel, and zinc (Machado et al. 2010b). The difficulty of the selective recovery of these four metals is well illustrated in a recent work published by Kuchar et al. (2010). In this paper, the sulfidation was conducted to a selective recovery of Cu and Zn (96.6% and 91.5%, respectively); however, only 64.3% and 30% of Ni and Cr, respectively, were selectively recovered.

In the present work, a multistage process, which combines electrochemical and chemical methods, was developed for

recovering selective and efficiently chromium, copper, nickel, and zinc from an acid solution. The acid solution resulted from the incineration of contaminated brewing cells of *S. cerevisiae* loaded with heavy metals. It is important to point out that the challenge presented in this work is more ambitious and difficult than the other one previously carried out (Machado et al. 2010d) since it implies the selective recovery of four metals; one of them [chromium(III)] requires a completely different strategy. With the strategy developed in this work, chromium, copper, nickel, and zinc were selectively recovered with high yield ( $\geq 83\%$  for all four metals). As far as we know, this is the first work, which reports the recovery of these four metals with high selectivity.

## 2 Materials and methods

### 2.1 Incineration of yeast cells and acid digestion of ashes

Brewing cells of *S. cerevisiae*, used in the bioremediation of an electroplating effluent, were dried at 45°C and then incinerated at 550°C, in a muffle furnace (Nabertherm, L5/C6), until constant weight. Then, the ashes were acid digested with hydrochloric acid 1 mol/L (prepared from proanalysis acid, Merck) as previously described (Machado et al. 2010d).

After the digestion of the ashes, total chromium, copper, nickel, and zinc concentrations were determined in the acid solution by atomic absorption spectroscopy with flame atomization (AAS-FA) in a Perkin Elmer AAnalyst 400 spectrometer (Norwalk, CT, USA) in order to control the efficiency of the digestion.

This experiment was performed in duplicate.

### 2.2 Copper recovery by electrolysis

Copper recovery was performed in a batch reactor, as previously described (Machado et al. 2010d). Briefly, an electrochemical cell, constituted by a platinum net as cathode, a platinum wire as anode and an electrode of Ag/AgCl (KCl 3 mol/L) as reference electrode, was immersed in 40 mL of the acid digestion solution containing chromium(III), copper, nickel, and zinc, previously adjusted at pH 2. Then, a constant cathodic potential of  $-0.6$  V was applied at 40°C with a constant agitation of 500 rpm. During copper electrodeposition, electric current intensity varied between 0.1 and 0.01 A.

In order to determine the efficiency of metal recovery, copper concentration was quantified in the remained acid solution by AAS-FA. For controlling the selective recovery of copper, total chromium, nickel, and zinc were quantified in the metallic copper, by AAS-FA, after redissolving it in nitric acid.

All experiments and determinations were done at least in duplicate.

### 2.3 Computer chemical simulations

Chemical speciation calculations were carried out using the MINEQL+ software (version 4.5; Schecher and McAvoy 2003). Metal speciation analysis with MINEQL+ generates chemical equilibrium concentrations of all species being considered in the model by the program reactions, based on component stability constants and molar concentrations. Computational simulations were performed considering the total heavy metal (chromium, nickel, and zinc quantified in the remained acid digested solution after copper deposition) concentrations and the solubility product constants (Martell and Smith 2004).

### 2.4 Nickel recovery by simultaneous alkalization at pH 14 and oxidation of chromium(III) to chromium(VI)

Since the final pH was very high (pH 14), the mass of sodium hydroxide necessary to adjust the pH was theoretically calculated taken into account the amount of hydroxide necessary to raise the pH from 2 to 14 plus the amount of hydroxide necessary to convert chromium, nickel, and zinc to chromate [ $\text{CrO}_4^{2-}(\text{aq})$ ], nickel hydroxide [ $\text{Ni}(\text{OH})_2(\text{s})$ ], and tetrahydroxozincate(II) [ $\text{Zn}(\text{OH})_4^{2-}(\text{aq})$ ], respectively; due to the high ionic strength of the solution, calculation of pH was performed considering the  $\text{OH}^-$  activity instead of  $\text{OH}^-$  concentration. Then, small pellets of solid sodium hydroxide were added. After all sodium hydroxide was dissolved, an excess of hydrogen peroxide [five times more the amount of chromium(III)] was added. Finally, the suspension was boiled during 15 min, in order to decompose the excess of hydrogen peroxide, and centrifuged at 2,800g for 10 min, for recovering the precipitate of nickel hydroxide,  $\text{Ni}(\text{OH})_2(\text{s})$ . Nickel hydroxide was firstly washed two times with 10 mL of NaOH at pH 14, followed by four washing steps with 10 mL of NaOH at pH 10; the precipitate was recovered by centrifugation at 2,800g during 10 min.

Oxidation of chromium(III) to chromium(VI) was controlled by determining the amount of total chromium and chromium(VI) in the remained supernatant solution. Total chromium and chromium(VI) were quantified by AAS-FA and the diphenylcarbazide method according to the standard methods (APHA et al. 1998), respectively.

For characterization of nickel hydroxide and determination of nickel recovery, concentrations of total chromium, copper, nickel, sodium, and zinc were measured in solution after appropriate dissolution in  $\text{HNO}_3$  20% (v/v) and dilution of a dried (at 105°C until constant weight) weighted amount of nickel hydroxide. Total chromium,

copper, nickel, and zinc were measured by AAS-FA. Sodium concentration was determined by emission atomic spectroscopy (EAS) in a Perkin Elmer AAnalyst 400 spectrometer (Norwalk, CT, USA). The amount of chlorides present in the precipitate was calculated as the difference between the total amount of a dried (at 105°C until constant weight) precipitate weighted gravimetrically and the sum of copper, nickel, zinc (all in the form of metal hydroxide), and sodium quantified in the precipitate.

All experiments and determinations were done at least in duplicate.

### 2.5 Zinc and chromium recovery by adjustment of pH at 10

After nickel recovery, the remained supernatant solution, containing chromium(VI) and zinc, was adjusted at pH 10 by addition of HCl 37% followed by centrifugation at 2,800g for 10 min. Zinc was recovered as a precipitate of zinc hydroxide,  $\text{Zn}(\text{OH})_2(\text{s})$ , and chromium remained in solution as chromate,  $\text{CrO}_4^{2-}(\text{aq})$ .

The amount of zinc recovered was determined by AAS-FA, after appropriate dissolution of part of the dried (at 105°C until constant weight) precipitate in  $\text{HNO}_3$  20% (v/v) and dilution.

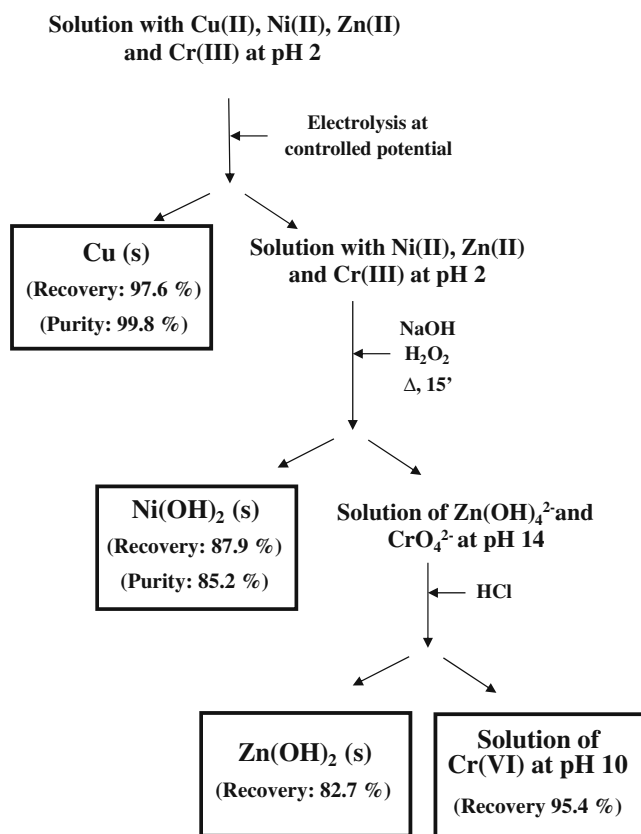
For characterization of chromate solution and quantification of chromium recovery, total chromium, copper, nickel, sodium, and zinc were determined in the remained supernatant solution as it was described above. Chlorides were analyzed according to the standard methods (APHA et al. 1998) by the argentometric method. All experiments and determinations were done at least in duplicate.

## 3 Results and discussion

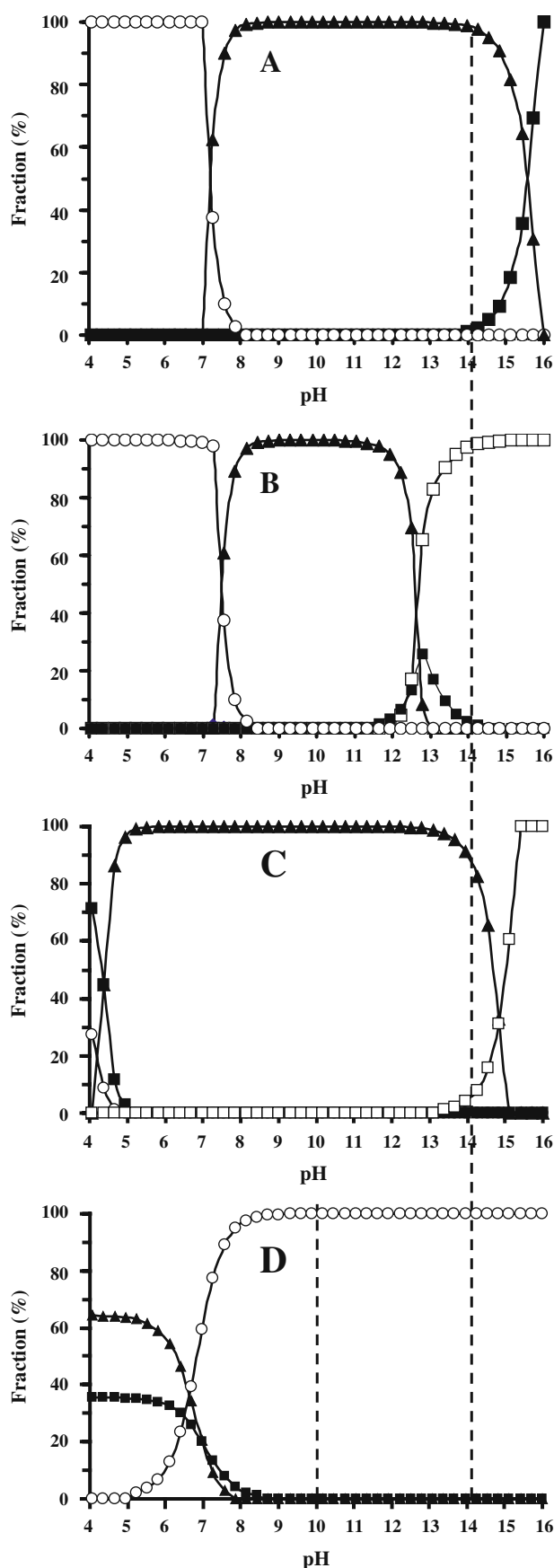
Brewing flocculent cells of *S. cerevisiae* were used to treat a real electroplating effluent containing chromium, copper, nickel, and zinc (Machado et al. 2010b). This biomass, loaded with heavy metals, was incinerated and subsequently acid digested under the conditions previously optimized (Machado et al. 2010d). The acid solution contained 3.8, 0.4, 2.8, and 0.2 g/L of chromium(III), copper, nickel, and zinc, respectively. This solution was used to develop a nearly closed cycle for recovering selectively chromium, copper, nickel, and zinc (Fig. 1). This work is a more ambitious and difficult challenge than the other one previously carried out (Machado et al. 2010d) since it implies the selective recovery of four metals; one of them [chromium(III)] requires a completely different strategy. If the methodology developed by us in the previous paper (Machado et al. 2010d) has been applied to the present case [an acid solution containing chromium(III), copper, nickel, and zinc], only the selective recovery of copper, by electrolysis, and of zinc, as a soluble complex of tetrahydroxozincate after the subsequent

alkalinization of the solution up to pH 14, would have been attained; nickel and chromium(III) would remain both precipitated, as nickel hydroxide and chromium(III) hydroxide (Fig. 2), respectively.

In the present work, copper was, firstly, electrolysed from the multimetal acid solution after previous pH adjustment of the acid digested solution at pH 2. After 1 h of electrolysis, a residual concentration of copper, 9 mg/L, remained in the acid solution. This corresponded to a recovery of 97.6% of copper (Fig. 1), which is in agreement with our previous results (Machado et al. 2010d) and the results from other authors. Suzuki et al. (1995) reported the complete removal of copper from a mixture of copper, nickel, and zinc by electrolysis at a controlled potential. In the same way, Aydin et al. (1998) reported the electrodeposition of 99.3% of copper from a solution containing bismuth, cadmium, cobalt, copper, nickel, and zinc. In the present work, the purity of the metal recovered was 99.8% (Fig. 1), containing minor amounts of chromium, nickel, and zinc (Table 1). Doulikas et al. (2000) had reported a similar copper purity (higher than 99.5%) when copper was recovered from a multielement solution containing cadmium, copper, lead, and zinc ions, at a controlled potential. The high purity of the final product



**Fig. 1** Diagrammatic representation of the overall environmentally friendly process proposed in the present work. The process allows an efficient and selective recovery of chromium, copper, nickel, and zinc from an acid solution



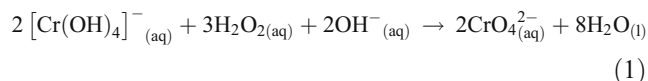
◀ **Fig. 2** Species distribution diagram of 2.8 g/L of nickel (a), 0.2 g/L of zinc (b), and 3.8 g/L of chromium(III) (c) or chromium(VI) (d), at different pH values. The different species of metals in solution were calculated with a chemical equilibrium computer program (MINEQL+) and assuming the total concentrations of metals remained in the acid digested solution after copper electrolysis. Nickel or zinc species: free metal ion,  $M_{(aq)}^{2+}$  (empty circle),  $M(OH)_{2(s)}$  (filled triangle),  $M(OH)_3^-(aq)$  (filled square) and  $M(OH)_4^{2-}(aq)$  (empty square). Chromium(III) species:  $Cr_{(aq)}^{3+}$  (empty circle),  $Cr(OH)_{(aq)}^{2+}$  (filled square),  $Cr(OH)_{3(s)}$  (filled triangle), and  $Cr(OH)_4^-(aq)$  (empty square). Chromium(VI) species:  $CrO_4^{2-}(aq)$  (empty circle),  $Cr_2O_7^{2-}(aq)$  (filled triangle), and  $HCrO_4^-(aq)$  (filled square)

obtained allows its introduction in the market. On the other hand, it can also be introduced in several types of copper plating baths (namely in sulfate copper acid baths and fluoborate acid baths as well as in copper pyrophosphate-plating baths and brass plating solutions; Homer 1994).

Then, the possibility of recovering selectively chromium, nickel, and zinc by chemical precipitation was evaluated. For this purpose, chemical speciation distribution diagrams, using the MINEQL+ program, were drawn assuming the concentrations of chromium, nickel, and zinc present in the remained acid solution (Fig. 2). The analysis of figure pointed out that at pH 6, chromium(III) should be totally precipitated, as  $Cr(OH)_3(s)$  (Fig. 2c), whereas nickel (Fig. 2a) and zinc (Fig. 2b) should remain soluble, as free metal ions, in solution. From a theoretical point of view, these results suggested that chromium(III) could be selectively recovered by adjusting the pH of the solution obtained after electrolysis at pH 6; nickel and zinc ions should remain in solution. However, experimental studies evidenced that part of nickel and zinc coprecipitated with chromium(III) hydroxide (data not shown), which unabled this strategy. On the other hand, the analysis of Fig. 2a, c evidenced that whatever the pH at which the solution was adjusted, at least two metals would be remained in the same phase (precipitated or soluble).

From these results, we definitively concluded that a new strategy should be developed to recover selectively the three metals. The analysis of the chemical speciation distribution diagram of chromium(VI) (Fig. 2d), together with the chemical speciation distribution diagrams for nickel (Fig. 2a) and zinc (Fig. 2b), pointed out that nickel could be selectively recovered, as a precipitate of nickel hydroxide, from the remaining solution [chromium and zinc will remain totally soluble as anions of chromate,  $CrO_4^{2-}$

$^-(aq)$ , and tetrahydroxozincate,  $Zn(OH)_4^{2-}(aq)$ , respectively] at pH 14. In a recent paper published by Kuchar et al. (2010), only 64.3% of nickel and 30% of chromium were selectively recovered from a plating sludge, containing chromium(III), copper, nickel, and zinc, after previous oxidation of chromium(III) to chromium(VI) at pH 10; in this paper (Kuchar et al. 2010), less than 60% of the total amount of chromium(III) was oxidized to chromium(VI). However, it is described in the literature (Vogel 1987) that oxidation of chromium(III) to chromium(VI) can be easily promoted using an excess of hydrogen peroxide, under alkaline conditions, according to the equation below:



The analysis of the chemical speciation distribution diagram of chromium(III) (Fig. 2c) evidences that at pH 14, about 5% of chromium(III) is solubilized as tetrahydroxochromate(III),  $[Cr(OH)_4]_{(aq)}^-$ . All these facts allowed us to design another strategy to recover selectively chromium and zinc. This strategy was based on the oxidation of chromium(III) to chromium(VI) under stronger alkaline conditions. Thus, we decided to increase the pH of the suspension up to 14 and add an excess of hydrogen peroxide (Fig. 1). Under these conditions, 95.4% of chromium(III) was rapidly (15 min were used) oxidized to chromium(VI), and 87.9% of nickel was recovered, as nickel hydroxide, with a purity of 85.2% (Fig. 1). Chromium and chlorides were the main contaminants present in the nickel hydroxide (Table 1). Minor amounts of copper, sodium, and zinc were also detected. The nickel hydroxide, obtained under the present conditions, can be reused to adjust nickel concentration in the Watts baths, which present a concentration of 30–60 g/L of  $NiCl_2 \cdot 6H_2O$  (Homer 1994). If a high purified final product is required, the recovered nickel hydroxide can be washed using NaOH at pH 10 in a similar way as it was done in a previous work (Machado et al. 2010d).

After nickel recovery, the remained solution was mainly constituted by high amounts of chromium(VI) and small amounts of zinc. Chemical speciation distribution diagrams of zinc (Fig. 2b) and chromium(VI) (Fig. 2d) predicted that, at pH 10, zinc should be totally precipitated, as  $Zn(OH)_2(s)$ , and chromium(VI) will remain in the supernatant, as  $CrO_4^{2-}(aq)$ .

**Table 1** Main impurities presented in metallic copper, nickel hydroxide, and chromate solution

Metals	Contaminants (% w/w)					
	Na	Cl	Cu	Ni	Zn	Cr
Cu(s)	—	—	—	0.050	0.060	0.070
Ni(OH) <sub>2</sub> (s)	0.30	11.4	0.40	—	0.40	2.3
CrO <sub>4</sub> <sup>2-</sup> (aq)	5.78	18.2	≤3.0×10 <sup>-5</sup>	≤3.0×10 <sup>-5</sup>	≤4.0×10 <sup>-5</sup>	—



So, the pH of the remained solution was adjusted to pH 10 with concentrated hydrochloric acid. After this procedure, 82.7% of the zinc initially present in the solution was recovered (Fig. 1). Since the initial solution only contained 0.2 g/L of zinc, the final amount of zinc hydroxide recovered was very low. So, no attempts to purify this amount of zinc hydroxide were performed; in this case, the recovered zinc hydroxide should be reintroduced in a subsequent initial acid digested solution. On the other hand, if significant amounts of zinc are present in the initial acid solution, large amounts of zinc hydroxide will be recovered. In this case, if a high purified final product is required, the recovered zinc hydroxide can be washed using NaOH at pH 10 in a similar way as it was done previously (Machado et al. 2010d).

The concentration of chromium present in the recovered cromate solution was 2.6 g/L, which corresponded to a recovery of 95.4% (Fig. 1). This solution contained chlorides and sodium ions, as major contaminants, and vestigial amounts of copper, nickel, and zinc (Table 1). This solution can be reused in the preparation of chromium plating solution of conventional (150–400 g/L of  $\text{CrO}_3$ ) and cocatalyzed baths (150–400 g/L of  $\text{CrO}_3$ ; Homer 1994).

A preliminary economic study was carried out to estimate the economic feasibility of the flowsheet proposed. Estimation of the cost was based on the following economic factors: (1) the energy consumption (and the cost) for copper electrodeposition; (2) the cost associated to the amount of sodium hydroxide consumed to increase the pH of the acid solution to 14; (3) the cost associated to the use of the hydrogen peroxide to oxidize chromium(III) to chromium(VI); and (4) the cost to decrease the pH of solution to 10. The preliminary cost analysis of the process, considering 1 m<sup>3</sup> of acid solution, is presented in Table 2. The prices of the end products were compared with the market prices of equivalent products. Unless copper, which can be recovered at a low cost, the other metals are recovered with a higher price than if they are purchased in the market. Since KOH (0.30 US\$/kg) is cheaper than

NaOH (0.50 US\$/kg; Chang 2011), the replacement of NaOH by KOH will reduce the total cost associated with the selective recovery of the metals from ~290 to ~228 US \$. Considering the use of KOH, the preliminary economic study pointed out that the end products are prepared with a cost of about two times higher than the market prices of equivalent products.

## 4 Conclusions

In this work, a new and multistep process for recovering selectively chromium, copper, nickel, and zinc, with high yield, from an acid solution was developed. For this purpose, a solution obtained after microwave acid (hydrochloric acid was used) digestion of the ashes produced from the incineration of contaminated biomass used to treat electroplating effluents, was used. The acid solution contained 3.8, 0.4, 2.8, and 0.2 g/L of chromium(III), copper, nickel, and zinc, respectively.

The developed process comprised three steps: (1) selective recovery of copper, as metallic copper, by electrolysis at controlled potential; (2) selective recovery of nickel, as nickel hydroxide, by simultaneous alkalization at pH 14 and oxidation of chromium(III) to chromium(VI) of the previous solution; (3) selective recovery of zinc, as zinc hydroxide, and of chromium(VI), as a solution of cromate, by adjusting the pH of the previous solution at pH 10. With this process, final recoveries of 95.4%, 97.6%, 87.9%, and 82.7% of chromium(VI), copper, nickel, and zinc, respectively, were obtained. The purity of the metals recovered allows reintroducing them in the electroplating process.

The process developed in the present work, at laboratory scale, seems to be a promissory approach to be applied in the selective recovery of metals present in an acid digestion solution resulted from the incineration of ashes of biomass used in the treatment of effluents containing four heavy

**Table 2** Preliminary cost analysis of the selective recovery of chromium (VI), copper, nickel, and zinc from an acid solution (1 m<sup>3</sup>)

Item	Unit	Quantity	US\$/unit	Cost
Energy for copper recovery	kWh	5.2	0.1	0.52
Amount of NaOH <sup>a</sup>	kg	310	0.5	155
Amount of H <sub>2</sub> O <sub>2</sub> (50%) <sup>a</sup>	kg	38.7	0.8	31
Amount of HCl (33%) <sup>a</sup>	kg	738.8	0.14	103.43
Total cost				289.95
Products				
Chromium <sup>b</sup>	kg	3.6	9.8	35.3
Copper <sup>c</sup>	kg	0.39	9.6	3.74
Nickel <sup>b</sup>	kg	2.5	30	75
Zinc <sup>b</sup>	kg	0.17	7.3	1.24
Total cost for products				115.28

<sup>a</sup> International Chemical Information Service (Chang 2011)

<sup>b</sup> Local market as a salt

<sup>c</sup> London Metal Exchange (2011)

metals. Although the price of the recovered metals is, globally, higher than if they are purchased from the market, the strategy proposed is environmentally friendly and can contribute to avoid the world mineral reserve exhaustion.

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