



# Leaching Kinetics of Selenium, Tellurium and Silver from Copper Anode Slime by Sulfuric Acid Leaching in the Presence of Manganese(IV) Oxide and Graphite<sup>+</sup>

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Abstract: Sulfuric acid leaching of copper anode slime/CAS in the presence of manganese(IV) oxide (MnO<sub>2</sub>) and graphite has been investigated for Se, Te and Ag recovery. The study reveals that the leaching of Se, Te and Ag was facilitated by the galvanic interaction with MnO<sub>2</sub>, and graphite played the role of a catalyst. The leaching process could yield 81.9% Se, 90.8% Te, and 80.7% Ag leaching efficiency when the conditions were maintained as: 500 rpm, 2.0 M H<sub>2</sub>SO<sub>4</sub>, 0.8/0.8/1 MnO<sub>2</sub>/graph-ite/CAS, and 90 °C temperature. The kinetic study showed that Se leaching followed the surface chemical reaction at all the tested temperature range (25–90 °C) with the activation energy of 27.7 kJ/mol. Te and Ag leaching at temperature 25–50 °C followed the mixed and surface chemical reaction models, respectively, and changed to fit the diffusion and mixed control models, respectively in the temperature range 60–90 °C with the corresponding activation energy of 17.8 and 12.2 kJ/mol.

Keywords: copper anode slime; tellurium; selenium; silver; galvanic leaching; kinetics

## 1. Introduction

Copper anode slime (CAS), a byproduct of copper electrorefining process, has long been used as a secondary resource for various critical metals, such as Au, Ag, Pt, Pd, Se, Te, Cu, etc. Metallurgical treatments of CAS for recovering these valuable metals include the stepwise thermal-aqueous, aqueous-thermal, aqueous-aqeuous processes, consisting of roasting to recover Se and oxidative leaching to Te and Cu, prior to the recovery of precious metals [1].Despite the extensive investigation on the aforesaid processes, problematic facts remain. For an instance, roasting of CAS either through oxidative or sulfation roasting results in loss of Se fraction. Another option of soda roasting is unfavorable in practice because of the subsequent acid leaching process which increases the reagent consumption for acidification/neutralization and produces low-cost byproducts. The similar problem appears when approaching alkaline pressure leaching for recovering selenium.

Recently, researches on the treatment of CAS have been focused to shorten the number of treatment stages to a single step. Sulfuric acid leaching in the presence of strong oxidant has shown the capability of simultaneous recovery of Se, Te, and even Ag. HNO<sub>3</sub> was used as the oxidant for the leaching of Se from decopperized CAS, which achieved 97.8% of Se recovery [2]; however, the process faces issue of the release of harmful NOx gas which required a complicated sealed-leaching system. H<sub>2</sub>O<sub>2</sub> is less toxic oxidant but its stability is sensitive to the higher temperature (starts to decompose at temperature  $\geq$  70 °C) and is unrecyclable as well [3]. MnO<sub>2</sub> provides low environmental impact, high metal recovery, and recyclability; according to Xiao et al. [4], sulfuric acid leaching with MnO<sub>2</sub> can ensure a profit of \$82 per 1 t of CAS treated. With this potential, the leaching

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**Copyright:** © 2021 by the authors. Submitted for possible open access publication under the terms and conditions of the Creative Commons Attribution (CC BY) license (http://creativecommons.org/licenses/by/4.0/). using MnO<sub>2</sub> critically depends on the constituent phases of the CAS, particularly selenides/tellurides.

In this study, leaching of constituent metals from CAS (Se, Te, Ag) has been investigated using H<sub>2</sub>SO<sub>4</sub> as the lixiviant, MnO<sub>2</sub> as the oxidant and graphite as the catalyst. The leaching conditions were optimized, and leaching kinetics of the targeted metals were studied while adopting the shrinking core models.

### 2. Materials and Methods

The CAS sample used was supplied by a copper smelter in South Korea. The wetchemical analysis of the sample revealed that the CAS contained 22.23% Se, 12.13% Ba, 9.66% Ag, 1.53% Te, and precious metals of 0.046% Au, 37.93 ppm Pd and 2.94 ppm Pt. The X-ray diffraction analysis (pattern given in **Fig. 1**) shows the presence of Se<sup>0</sup>, Ag<sub>2</sub>Se and BaSO<sub>4</sub> phases. For the leaching studies, a known concentration of H<sub>2</sub>SO<sub>4</sub> solution was put into a glass reactor fitted over a water bath. Once the desired temperature was achieved, a weighed amount of CAS was added to the lixiviant while maintaining the solid/liquid ratio of 2.5 g/250 ml of H<sub>2</sub>SO<sub>4</sub> solution. The leaching was performed at the stirring speed of 500 rpm for 6 h (after preliminary studies), and slurry liquid samples were withdrawn at regular times, filtered and diluted appropriately into 5% nitric acid for analyzing metal contents using inductively coupled plasma spectroscopy (ICP). Chemicals such as H<sub>2</sub>SO<sub>4</sub> (98%, Junsei Chemical, Japan), MnO<sub>2</sub> (87%, Wako Chemical, Japan) and graphite (Kanto Chemical, Japan) were used as received.

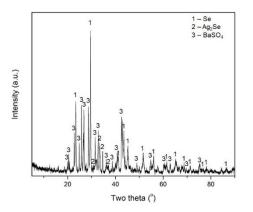
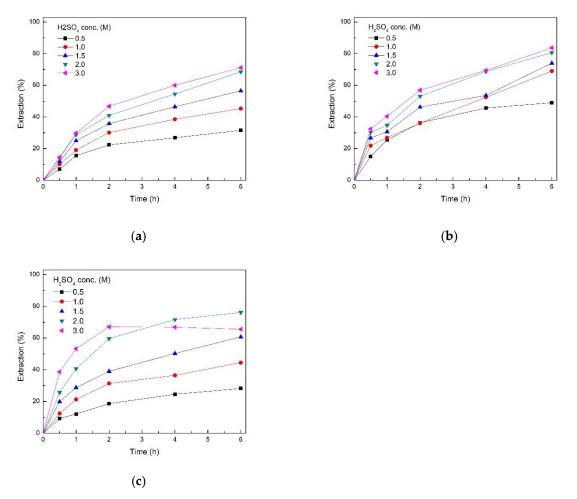


Figure 1. XRD pattern of raw copper anode slime

#### 3. Results

#### 3.1. Effect of H<sub>2</sub>SO<sub>4</sub> concentration

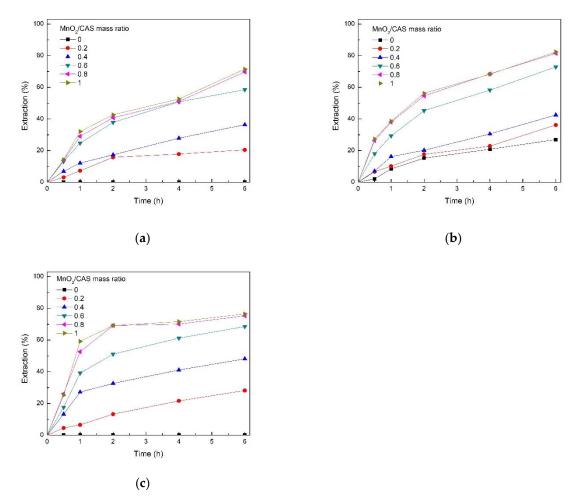
The effect of H<sub>2</sub>SO<sub>4</sub> in the range 0.5–3.0 M on the extraction percentage was investigated. For this, leaching conditions were maintained as: stirring speed 500 rpm; MnO<sub>2</sub>/graphite/CAS mass ratio 0.8/0.8/1, and temperature 70 °C. The extraction behavior of Se, Te and Ag as a function of time at the investigated H<sub>2</sub>SO<sub>4</sub> concentrations is given in **Fig. 3**. As can be seen the leaching of Se, Te and Ag increased significantly by increasing H<sub>2</sub>SO<sub>4</sub> from 0.5 to 2.0 M; at 2.0 M H<sub>2</sub>SO<sub>4</sub>, the leaching percentages of the metals are 80.7%, 68.6% and 76.8%, respective. At still higher H<sub>2</sub>SO<sub>4</sub> concentration, the leaching trends of Se and Te could show marginal improvements but that of Ag decreased which is attributed to the formation of Ag<sub>2</sub>S [6–8] to limit the leaching of Ag-related phases [3–5]. Therefore, the further leaching experiments were performed using 2.0 M H<sub>2</sub>SO<sub>4</sub>.



**Figure 2.** Dissolution behavior of (a) selenium, (b) tellurium and (c) silver in H<sub>2</sub>SO<sub>4</sub> solution as a function of time (variation of H<sub>2</sub>SO<sub>4</sub> conc., 0.5–3.0 M;stirring speed 500 rpm; MnO<sub>2</sub>/graphite/CAS mass ratio 0.8/0.8/1; and temperature 70 °C)

## 3.2. Effect of MnO<sub>2</sub> dosage

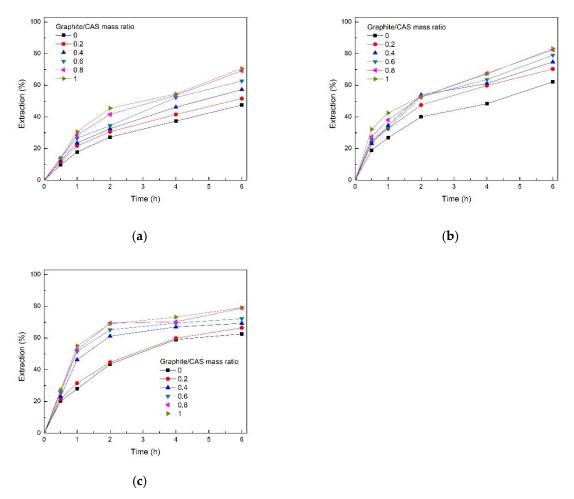
Here, MnO<sub>2</sub> dosage was expressed as the mass ratio of MnO<sub>2</sub> to CAS. Experiments were conducted under the conditions mentioned in **Fig. 3**. It can be seen that MnO<sub>2</sub> has pronounced effect on the leaching of Se, Te and Ag. In the absence of MnO<sub>2</sub>, leaching yield of Te was 32.8% and those of Se and Ag were negligible. Increasing MnO<sub>2</sub>/CAS mass ratio increased the leaching of the three metals; at MnO<sub>2</sub>/CAS mass ratio 0.8/1, the leaching of Se, Te and Ag was 81.5%, 69.8% and 75.3%, respectively. Further increase in MnO<sub>2</sub>/CAS mass ratio resulted in limited improvement which could be ascribed to the high crowding of metallic ions in the solution [9–12]. Thus, the MnO<sub>2</sub>/CAS mass ratio 0.8/1 was chosen for the next experiments.



**Figure 3.** Dissolution behavior of (a) Se, (b) Te and (c) Ag in 2.0 M H<sub>2</sub>SO<sub>4</sub> solution as a function of time (variation of MnO<sub>2</sub>/CAS mass ratio, 0–1;stirring speed 500 rpm; graphite/CAS mass ratio 0.8/1; and temperature 70 °C)

## 3.3. Effect of graphite dosage

**Fig. 4** shows the results of leaching experiments while varying the graphite/CAS mass ratio (ranging from 0 to 1). Experiments were performed under the conditions: stirring speed 500 rpm; H<sub>2</sub>SO<sub>4</sub> conc. 2.0 M; MnO<sub>2</sub>/CAS mass ratio 0.8/1, and temperature 70 °C. It can be seen that the leaching yields of Se, Te and Ag have increased from 40.2% to 69.2%, 54.5% to 82.4% and 56.6% to 78.9%, respectively, when the graphite/CAS mass ratio was increased from 0 to 0.8/1. More increasingly the graphite dosage did not improve the leaching yields because the graphite particles hindered the contact between the CAS particles and lixiviant [10–11]. Thus, graphite/CAS mass ratio of 0.8/1 was chosen for the subsequent experiments.



**Figure 4.** Dissolution behavior of (a) Se, (b) Te and (c) Ag in 2.0 M H<sub>2</sub>SO<sub>4</sub> solution as a function of time (variation of MnO<sub>2</sub>/CAS mass ratio, 0–1;stirring speed 500 rpm; MnO<sub>2</sub>/CAS mass ratio 0.8/1; and temperature 70 °C)

#### 3.4. Kinetic study

The kinetics of leaching reactions of Se, Te and Ag is described by the shrinking core model, wherein the reaction involves the reaction between solid particles and the fluid extractants. Two common equations of shrinking core model such as diffusion through a product layer (Eq. (1)) and surface chemical reaction (Eq. (2)) [3, 13–14], and an empirical mixed kinetic model (Eq. (3)) [15] were used to study the kinetics of leaching reactions.

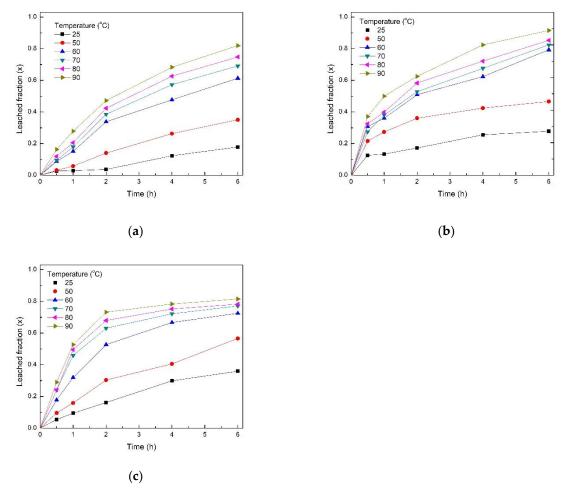
$$1 - \frac{2}{3}(1-x) + (1-x)^{\frac{2}{3}} = k_d t \tag{1}$$

$$1 - (1 - x)^{\frac{1}{3}} = k_r t \tag{2}$$

$$1 - (1 - x)^{\frac{1}{3}}]^2 = k_m t \tag{3}$$

where x is the fraction of metal leached in time t and k is the apparent rate constant with the corresponding subscript denoting the model type.

For the kinetic study, temperature of leaching was varied from 25 to 90 °C. Fig. 5 presents the leaching fraction of Se, Te and Ag as a function of time at the investigated temperatures, showing that increasing the temperature could increase the leached fraction of metals. At 90 °C, the leaching of Se, Te and Ag was 81.9%, 90.8% and 80.7%, respectively. This data is thus used for determining the rate constants and correlation coefficients based on the Eqs. (1–3), and details of which are provided in **Table 1**.



**Figure 5.** Leached fraction of (a) Se, (b) Te and (c) Ag in 2.0 M H<sub>2</sub>SO<sub>4</sub> solution as a function of time (variation of temperature, 25–90 °C; stirring speed 500 rpm; MnO<sub>2</sub>/CAS mass ratio 0.8/1)

	11							
Temperature (°C)	$1 - \frac{2}{3}(1 - x) + (1 - x)^{\frac{2}{3}} = k_d t$		$1 - (1 - x)^{\frac{1}{3}} = k_r t$		$[1 - (1 - x)^{\frac{1}{3}}]^2 = k_m t$			
	k <sub>d</sub>	R <sup>2</sup>	k <sub>r</sub>	R <sup>2</sup>	k <sub>m</sub>	R <sup>2</sup>		
	Se							
25	0.0006	0.92	0.0105	0.97	0.0007	0.91		
50	0.0028	0.96	0.023	0.99	0.0031	0.96		
60	0.0102	0.98	0.0442	0.98	0.0123	0.97		
70	0.0132	0.93	0.0519	0.99	0.0167	0.90		
80	0.0179	0.98	0.0621	0.99	0.0234	0.97		
90	0.023	0.99	0.0711	0.99	0.032	0.97		
	Te							
25	0.0017	0.97	0.0149	0.85	0.0018	0.98		
50	0.005	0.95	0.0262	0.78	0.0058	0.96		
60	0.0189	0.97	0.0584	0.92	0.0259	0.96		
70	0.0222	0.99	0.0662	0.96	0.0312	0.96		
80	0.0249	0.99	0.0708	0.95	0.0362	0.98		
90	0.03	0.99	0.0797	0.93	0.0481	0.98		
	Ag							
25	0.0031	0.98	0.0232	0.98	0.0034	0.98		

**Table 1.** Apparent rate constants  $k_d$ ,  $k_r$  and  $k_m$  for the kinetic models and correlation coefficients

50	0.08	0.96	0.0384	0.99	0.0095	0.95
60	0.0167	0.97	0.057	0.91	0.0218	0.98
70	0.0192	0.93	0.0617	0.79	0.0261	0.95
80	0.0202	0.88	0.0655	0.77	0.0278	0.91
90	0.0225	0.86	0.0598	0.83	0.032	0.89

For Se, the leaching rate at all investigated temperatures followed the surface chemical reaction with R<sup>2</sup>> 0.97. The apparent activation energy, calculated by Arrhenius equation ( $k = A. \exp(-\frac{Ea}{RT})$ ) and the plot of  $\ln(k)$  vs the reciprocal of the absolute temperature (1/*T*) (**Fig. 6**) was found to be 27.7 kJ/mol, which was lower than the typical range of chemically-controlled reaction (> 41.8 kJ/mol, [16]); this could be attributed to the catalytic effect of graphite to lower the activation energy [17–19].

For Te, change of rate-controlling step from mixed model (temperature 25–50 °C) to the diffusion control model (60–90 °C) was observed. This could be explained by the poor rate of chemical reaction at low temperature [16] while at high temperatures, the reaction becomes faster. Diffusion of leached Te species could happen through the Ag<sub>2</sub>S phase formed during the leaching as Te showed paragenitic relationship with Se in the Ag<sub>2</sub>Se phase. The calculated activation energy acquired during the diffusion controlled step (**Fig. 6**) was found to be 17.8 kJ/ mol in the temperature rage mentioned (60-90 °C).

Such a change in the leaching mechanism was also found in the case of Ag. The leaching was controlled by chemical reaction at temperatures 25–50 °C and turned to be controlled by the mixed controlled mechanism at the higher temperatures which could be attributed to the similar reason as that of Te. And the activation energy acquired in the latter case (mixed control) from the slope analysis of **Fig. 6** was 12.2 kJ/mol (60–90 °C); the low activation energy recorded was because of the catalytic effect of graphite [17–19].

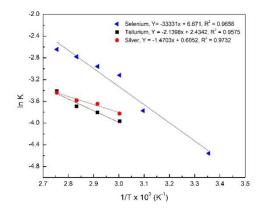


Figure 6. Arrhenius plot of Se, Te and Ag

#### 4. Conclusion

Se, Te and Ag from copper anode slime (CAS) were recovered by sulfuric acid leaching in the presence of MnO<sub>2</sub>. Graphite acted as the catalyst for the leaching reaction. Increasing leaching parameters, such as H<sub>2</sub>SO<sub>4</sub> concentration, MnO<sub>2</sub> and graphite dosage, and temperature resulted in significant improvements in the leaching of the three metals. With a stirring speed of 500 rpm, 81.9% Se, 90.8% Te and 80.7% Ag were leached out at H<sub>2</sub>SO<sub>4</sub> conc. 2.0 M, MnO<sub>2</sub>/graphite/CAS mass ratio 0.8/0.8/1 and temperature 90 °C after 6 h time. Kinetic investigation showed that the leaching of the three metals followed the shrinking core model. Se leaching proceeded through the surface chemical reaction with the apparent energy activation of 27.7 kJ/mol (25–90 °C). For Te and Ag, change in the rate-controlling steps from mixed control to the diffusion control and chemical control to mixed control were observed after increasing the temperature to  $\geq$  60 °C; the activation energy calculated being 17.8 and 12.2 kJ/mol (60–90 °C), respectively.

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