

Selective Leaching of LiFePO₄ by H₂SO₄ in the Presence of NaClO₃

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Abstract: Herein, problems commonly observed for the wet leaching of waste LiFePO₄ cathode materials, namely extensive Fe leaching and impurity removal, are mitigated through the use of NaClO₃, and the effects of leaching parameters on Fe, P, Li, and Al leaching efficiencies are probed. As a result, optimal leaching conditions are determined as temperature = 90° C, H_2 SO₄ concentration = 1 M, liquid-to-solid ratio = 5:1, leaching time = 1 h, stirring speed = 150 rpm, and NaClO₃ dosage = 25 g per 100 g raw material, with the corresponding Fe, P, Li, and Al leaching efficiencies obtained as 0.21, 0.03, 97.23 and 11.87%, respectively.

Keywords: spent lithium iron phosphate, selective leaching, cathode material, battery recycling

1. Introduction

Methods of spent LiFePO₄ battery disposal include the cascade utilisation and recovery of valuable metals [1,2]. Cascade utilisation aims to ensure the quality and safety of target products but is only applied in some demonstration plants, as the performance of waste batteries is unpredictable [3].

At present, the hydrometallurgical recovery of Li from waste LiFePO₄ batteries is widely employed in the lithium battery recovery industry [4-6]. Zhen et al. [7] leached the LiFePO₄ cathode material with a mixture of H₂SO₄ and H₂O₂ and adjusted the *p*H of the obtained solution with alkali to precipitate Fe as Fe(OH)₃, further adjusting *p*H to 5.0–8.0 to remove other heavy metal ions and thus obtain a solution of Li₂SO₄. The latter solution was treated with solid Na₂CO₃, concentrated, and crystallised to obtain Li₂CO₃. The above method consumes large amounts of alkali and produces much slag, therefore not complying with the concept of environmental protection and economy.

Yang et al. [8] probed the effects of mechanochemical activation of spent cathode powder and its potential to achieve selective Li recovery, revealing that after mechanochemical activation, ~97.67% Fe and 94.29% Li could be recovered under optimised conditions. The lixivium was evaporated and concentrated to recover FePO₄, and the filtrate *pH* was adjusted to neutral to afford Li₃PO₄. The above process is lengthy, featuring the drawbacks of high energy/alkali consumption and affording a product with high impurity content. Li et al. [9] found that the use of dilute H₂SO₄ as a leachant and H₂O₂ as an oxidant allows Li to be selectively leached into solution, while Fe and P remain in the residue as FePO₄, which is different from the traditional process of using excess mineral acid to leach all elements into solution.

Under optimised conditions, Li, Fe, and P leaching efficiencies of 96.85, 0.027, and 1.95%, respectively, were recorded. However, the above study is incomplete, employing an insufficient amount of experimental materials and therefore providing unconvincing results. Moreover, the chemical stability of H_2O_2 is low, the leaching process is prone to overflow, and the achieved oxidation efficiency is far lower than that observed for NaClO₃ and NaClO.

When waste LiFePO₄ cathode material is treated with H₂SO₄, Fe is solubilised as Fe²⁺, and its removal consumes large amounts of alkali liquor and produces much solid waste [10,11]. Importantly, the metal value of LiFePO₄ is lower than that of lithium nickel cobalt manganate, which results in low market value and low recovery enthusiasm [12–14]. Herein, LiFePO₄ was leached with H₂SO₄ in the

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presence of NaClO₃ at controlled pH and system potential to realise the removal of Fe in the form of insoluble FePO₄ to afford a solution of Li₂SO₄ and thus achieve selective Li leaching.

Thermodynamic analysis of the Li-Fe-P-H₂O system

During the leaching of LiFePO₄ cathode powder by H₂SO₄, pH and electric potential control are important for keeping Li in solution and Fe precipitated as FePO₄[1,12,13], which highlights the need to investigate the φ-pH diagram of the Li-Fe-P-H₂O system (Figure 1). According to the equilibrium reactions in E-pH Diagrams for the Li-Fe-P-H₂O System from 298 to 473 K [15], ion concentrations of 1 and 0.1 M were used instead of activities.

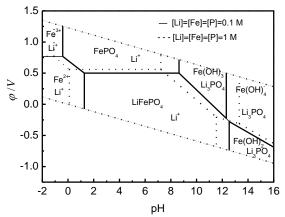


Figure 1. φ-pH diagram of the Li-Fe-P-H₂O system, T=298.15k

Figure 1 shows that the dominant area of FePO₄ in the Li-Fe-P-H₂O system is characterised by high potential and low pH. During the leaching of LiFePO₄ by acid, control of pH and system potential allows Fe and Li to be present as FePO₄ and Li⁺, respectively, thus enabling selective leaching and FePO₄ isolation.

In the absence of an oxidant, the reaction of LiFePO₄ with H₂SO₄ can be described as

$$LiFePO_4 + H_2SO_4 = Li_2SO_4 + FeSO_4 + H_3PO_4$$
 (1)

Conversely, a different reaction is observed in the presence of NaClO₃:

$$6LiFePO_4 + 3H_2SO_4 + NaClO_3 = 3Li_2SO_4 + 3FePO_4 \downarrow + NaCl + 3H_2O$$
(2)

2.Materials and methods

Concentrated H₂SO₄ and NaClO₃ were of analytical-grade purity. Spent LiFePO₄ was provided by Hunan Brunp Recycling Technology Co., Ltd., and contained 4.26 wt% Li, 17.39 wt% P, 32.18 wt% Fe, 2.13 wt% Al, and 44.03 wt% other elements.

A beaker filled with H₂SO₄ of a certain concentration and volume was charged with LiFePO₄ (100 g) and put into a water bath held at a specified temperature and equipped with an overhead mixer. Subsequently, NaClO₃ was added, and after a fixed retention time, the reaction mixture was filtered, and the Fe, P, Li and Al contents of the dry filter cake were determined. The filtration residue was analysed by diffraction of x-rays (XRD Ultima IV, JP) and scanning electron microscope (SEM Apreo C, US), The filtrate was analysed by inductively coupled plasma spectrometry (ICAP 7000SERIES, UK), and leaching efficiency (ηi) was calculated as

$$\eta_{i} = \frac{m_{A} - m_{a}}{m_{A}} \times 100\% \tag{3}$$



where m_A is the mass of a given element in the raw material, and m_a is the mass of this element in the filter residue.

3. Results and discussions

3.1. Effect of leaching temperature on leaching efficiency

Figure 2 shows the effects of temperature (T) on leaching efficiency at $[H_2SO_4] = 1$ M, liquid-to-solid ratio (L/S) = 5:1, leaching time (t) = 1 h, NaClO₃ dosage (D) = 25 g, and stirring speed (s) = 150 rpm.

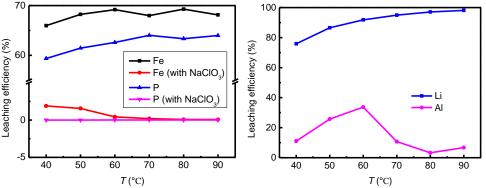


Figure 2. Effect of temperature on leaching efficiency

With increasing T, the leaching efficiency of Li increased, that of Fe declined, and that of P did not significantly change, and that of Al increased first and then decreased. At 90°C, the leaching efficiencies of Fe, P, Li, and Al equalled 0.07, 0.017, 98.12, and 16.72%, respectively. In order to achieve the results of high lithium leaching rate and low iron leaching rate, the above temperature (90 °C) was therefore selected as optimal.

3.2. Effect of sulphuric acid concentration on leaching efficiency

Figure 3 shows the effect of $[H_2SO_4]$ (T = 90 °C, L/S = 5:1, t = 1 h, D = 25 g, s = 150 rpm) on the leaching efficiencies of Fe, P, Li, and Al.

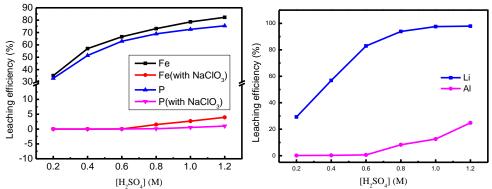


Figure 3. Effect of sulphuric acid concentration on leaching efficiency

With increasing $[H_2SO_4]$, the leaching efficiencies of Li and Fe increased, while that of P did not change significantly. At $[H_2SO_4] = 1$ M, the leaching efficiencies of Fe, P, Li, and Al reached 3.568, 0.547, 97.53, and 12.57%, respectively. When $[H_2SO_4]$ was further increased to 1.2 M, Li leaching efficiency did not significantly change, while Fe leaching efficiency linearly increased. Thus, the optimal $[H_2SO_4]$ value was concluded to be 1 M.



3.3. Effect of the liquid-to-solid ratio on leaching efficiency

Figure 4 shows the effect of L/S on the leaching efficiencies of Fe, P, Li, and Al ($T = 90^{\circ}$ C, $[H_2SO_4] = 1 \text{ M}$, t = 1 h, D = 25 g, s = 150 rpm).

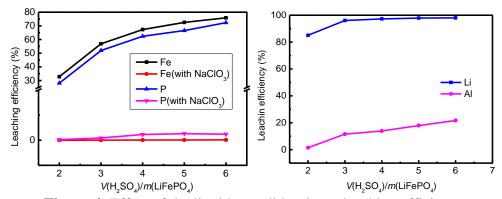


Figure 4. Effect of the liquid-to-solid ratio on leaching efficiency

An increase of L/S can reduce solution viscosity and thus accelerate mass transfer between solid and liquid phases to facilitate Li and Fe leaching. Hence, with increasing L/S, Li and Fe leaching efficiencies increased and then reached a plateau. At L/S = 5:1, the leaching efficiencies of Fe, P, Li, and Al equalled 0.26, 0.012, 97.896, and 17.93%, respectively. As these values stayed roughly constant at higher ratios, L/S = 5:1 was chosen as optimal.

3.4. Effect of leaching time on leaching efficiency

Figure 5 shows the effects of t on leaching efficiency (T = 90 °C, [H₂SO₄] = 1 M, L/S = 5:1, D = 25 g, s = 150 rpm).

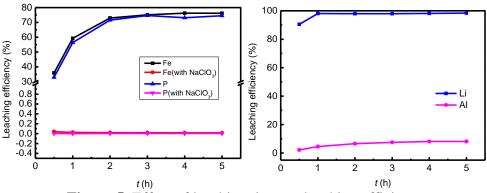


Figure 5. Effect of leaching time on leaching efficiency

With increasing t, the leaching efficiency of Li increased and then saturated, while those of Fe and P did not significantly change. After 1-h leaching, the leaching efficiencies of Fe, P, Li, and Al reached 0.025, 0.001, 98.099, and 4.54%, respectively. Thus, t = 1 h was chosen as optimal.

The leaching efficiency of P and Fe significantly increased under conventional leaching. In the presence of NaClO₃, all Fe²⁺ ions were oxidised to Fe³⁺ and precipitated together with P as FePO₄.

3.5. Effect of NaClO₃ dosage on leaching efficiency

Figure 6 shows the effects of *D* on leaching efficiency (T = 90 °C, [H₂SO₄] = 1 M, L/S = 5:1, t = 1 h, s = 150 rpm).



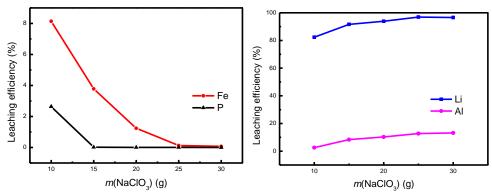


Figure 6. Effect of NaClO₃ dosage on leaching efficiency

With increasing D, the leaching efficiency of Li increased, while those of Fe and P decreased. At D = 25 g per 100 g raw material, Fe, P, Li, and Al leaching efficiencies reached 0.13, 0.004, 96.88, and 12.71%, respectively, and the above dosage was therefore chosen as optimal.

3.6. Effect of stirring speed on leaching efficiency

Figure 7 shows the effects of *s* on leaching efficiency (T = 90 °C, $[H_2SO_4] = 1$ M, L/S = 5:1, t = 1 h).

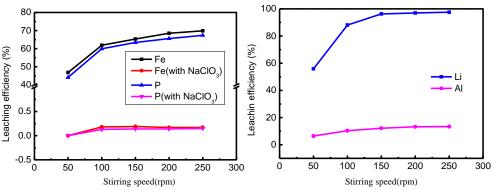
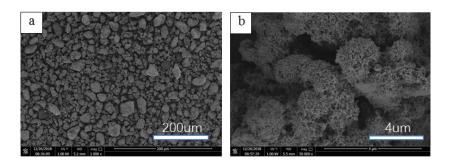


Figure 7. Effect of stirring speed on leaching efficiency

When *s* increased from 50 to 100 rpm, the leaching efficiencies of all elements increased. Upon a further increase to 150 rpm, P, Fe, and Al leaching efficiencies did not change significantly, while Li leaching efficiency increased to 96.26%, remaining stable at higher *s*. This behaviour was ascribed to the promotional effect of high stirring speed on substance diffusion.

3.7. Verification test

On the basis of single-factor experiments, optimum leaching conditions were determined as T = 90 °C, $[H_2SO_4] = 1$ M, L/S = 5:1, t = 1 h, D = 25 g per 100 g raw material. Under these conditions, the leaching efficiencies of Fe, P, Li, and Al equalled 0.21, 0.03, 97.23, and 11.87%, respectively.





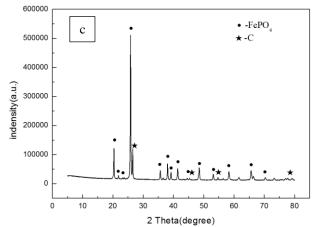


Figure 8. (a, b) SEM images and (c) XRD pattern of leaching residue

The morphological features of leaching residue particles obtained after leaching at optimum leaching conditions are displayed in Figure 8(a,b), It can be observed that the particles of leached residue are of different sizes (Figure 8a), and the surface of leached slag is of flocculent structure (Figure 8b). In order to identify the crystalline structure of leaching residue particles, XRD spectra were recorded and presented in Figure 8c. In order to identify the crystalline structure of leaching residue particles, XRD spectra were recorded and presented in Figure 8c. As can be seen from the above figures, LiFePO₄ was leached with H₂SO₄ in the presence of NaClO₃ at controlled *pH* and system potential, FePO₄ is the main component of the leaching residue. This indicates in the technology, we can removal of Fe in the form of insoluble FePO₄ to afford a solution of Li₂SO₄ and thus achieve selective Li leaching.

4. Conclusions

Herein, we realised selective leaching of Li from spent LiFePO₄ cathodes, showing that in the presence of NaClO₃, all Fe²⁺ ions in solution were oxidised to Fe³⁺ and precipitated together with P as FePO₄ to afford a P- and Fe-free solution containing Li.

- 1) Optimum conditions for the selective leaching of LiFePO₄ were determined as leaching temperature = 90° C, [H₂SO₄] = 1 M, L/S = 5:1, leaching time = 1 h, NaClO₃ dosage = 25 g per 100 g raw material.
- 2) FePO₄ can be obtained by selective leaching. After purification and modification, pure FePO₄ was obtained, which was used to prepare LiFePO₄ and realize resource recycling.

The paper, written in English, will publish on the website as .pdf file, ONE COLUMN; it should better between 4 and 12 pages. The article should be composed of the title, author(s), abstract, keywords, introduction, materials and methods, results and discussion, conclusions, and references.

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Manuscript received: 18.01.2020