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Efficient direct regeneration of spent LiFePO_4 from various degradation states for sustainable battery recycling†

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Efficient regeneration of spent LiFePO_4 is essential for the sustainable management of end-of-life electric vehicle batteries, given their significant market share. A universal method capable of directly regenerating spent LiFePO_4 from cells with varying states of health is crucial for practical implementation. Herein, we developed an oxalic acid/lithium hydroxide-based regeneration approach that effectively restores mixed spent LiFePO_4 , achieving a high specific capacity of $163.1 \text{ mA h g}^{-1}$ at 0.1C.

With the rapid expansion of the EV market, the volume of retired EV batteries has grown significantly in recent years, a trend expected to accelerate globally in the coming decades.^{1–3} Effective management of end-of-life (EoL) EV batteries is crucial not only for environmental sustainability but also for economic viability due to the high value and potential toxicity of battery materials.^{2,4–6} Battery recycling represents the final stage of a battery's lifecycle, though repurposing or remanufacturing may precede it.^{5,7} Recycling methods are generally classified based on their end products: if the process yields raw materials for lithium-ion battery (LIB) production, it is termed recovery;^{8–12} if the end product is active materials suitable for direct reuse in LIBs, it is termed direct regeneration.^{13–20}

Conventional recovery, primarily based on extracting key valuable materials (*e.g.*, Li, Co, Ni), has already been industrialized due to its operational simplicity.^{11,12,21} This method is particularly advantageous for lithium cobalt oxide (LCO)^{22,23} and nickel manganese cobalt (NMC) batteries.^{10,24,25} However, recovering lithium iron phosphate (LFP) batteries is less economically attractive, as the only recoverable high-value products are lithium-based salts (*e.g.*, lithium hydroxide or lithium carbonate). While lithium salts have higher unit mass values

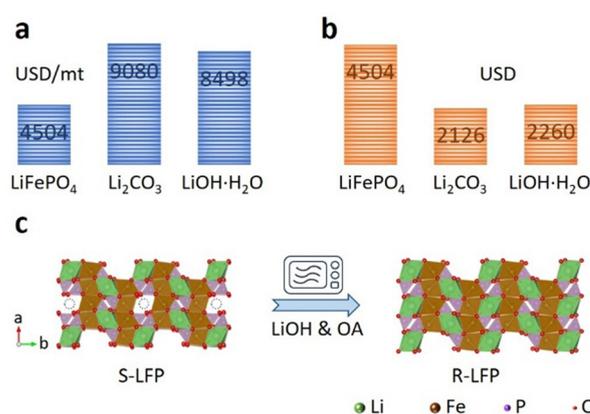


Fig. 1 (a) Market price of battery-grade LiFePO_4 , Li_2CO_3 , and $\text{LiOH}\cdot\text{H}_2\text{O}$ (unit: US dollars per metric ton) and (b) end-of-product price of recycling 1 ton of spent LiFePO_4 (assuming 100% conversion). (c) Illustration of recovery mechanism of S-LFP. (R-LFP denotes regenerated LFP).

than LFP (Fig. 1(a)), direct regeneration of 1 ton of spent LFP (denoted as S-LFP) batteries can double the product value (Fig. 1(b) and Table S1, ESI†), making this approach significantly more viable (Fig. S1, ESI†).

Existing direct regeneration strategies for LFP batteries primarily focus on replenishing lithium deficiencies in spent LFP.^{19,26} These methods typically utilize a reductant—often an organic acid, such as citric acid,²⁷ L-ascorbic acid,^{18,28,29} or tartaric acid,^{16,17} along with a lithium source, typically lithium hydroxide. While these methods have shown promise, their application remains largely confined to laboratory-scale studies, with several factors limiting their practical feasibility. One key challenge is whether a given method is suitable for mixed spent LFP derived from different battery sources. To address this issue, we developed a rapid microwave-assisted hydrothermal method (Table S2, ESI†) utilizing oxalic acid (OA) as a reducing and chelating agent, along with lithium hydroxide as the lithium source. This method was applied to mixed spent LFP from various batteries with different states of health (SOH). Additionally, we

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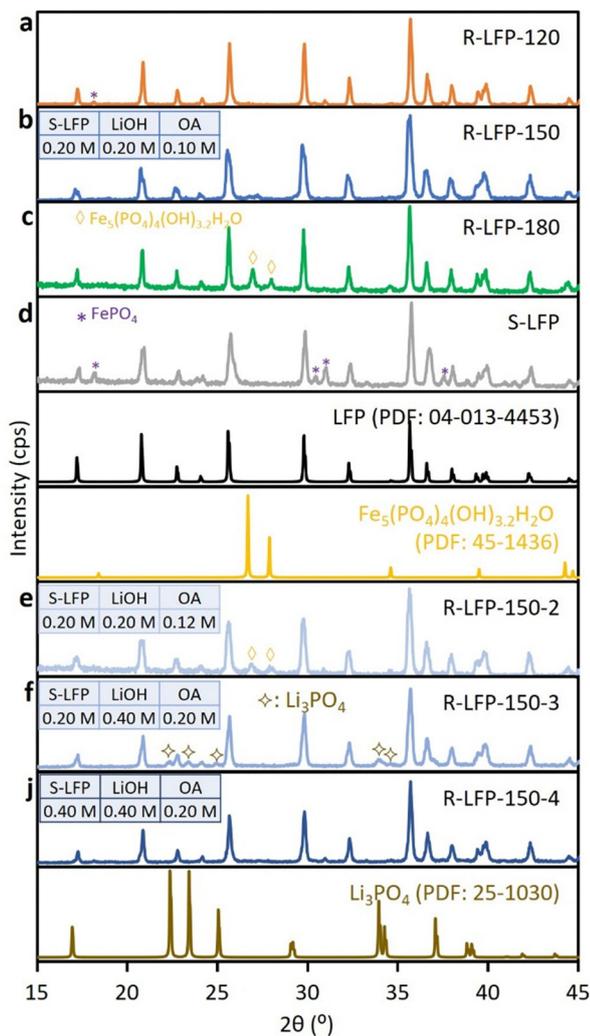


Fig. 2 XRD Patterns of R-LFP-120 (a), R-LFP-150 (b), R-LFP-180 (c), S-LFP (d), R-LFP-150-2 (e), R-LFP-150-3 (f) and R-LFP-150-4 (g). 120, 150, 180 indicate the reaction temperature ($^{\circ}\text{C}$). R-LFP denotes regenerated LFP.

systematically investigated the effects of reaction temperature and the molar ratios of OA, lithium hydroxide (LiOH), and spent LFP on the regeneration efficiency.

We cycled 1.5 A h LFP cylindrical cells to different states of health (SOH), ranging from 55% to 70% (Fig. S2, ESI †). The spent LFP (S-LFP) was obtained from these cells and mixed together. The separation and purification processes are detailed in the ESI † . The XRD pattern of S-LFP (Fig. 2(d)) exhibits specific peaks similar to reported result,²⁷ with impurity peaks of FePO_4 appearing around 18 degrees.

We first examined the impact of reaction temperature on direct LFP regeneration. XRD analysis (Fig. 2(a)–(c)) revealed that excessively high temperatures (180 $^{\circ}\text{C}$) led to the formation of $\text{Fe}_5(\text{PO}_4)_4(\text{OH})_3 \cdot 2\text{H}_2\text{O}$ (PDF: 45-1436), an undesirable impurity. Meanwhile, FePO_4 residues persisted in R-LFP-120, likely due to the lower reaction rate at 120 $^{\circ}\text{C}$. The optimal temperature was 150 $^{\circ}\text{C}$, as R-LFP-150 exhibited the best match with simulated LFP (PDF: 04-013-4453).

Based on literature findings, organic acids play a critical role in the direct regeneration of spent LFP, acting as reducing agent.^{16,18,27} To assess the impact of increasing OA concentration, we added 20 wt% more OA to the reactor while maintaining constant amounts of S-LFP and LiOH. The XRD pattern of R-LFP-150-2 (Fig. 2(e)) indicated that excessive OA resulted in the formation of the $\text{Fe}_5(\text{PO}_4)_4(\text{OH})_3 \cdot 2\text{H}_2\text{O}$ phase, similar to that observed in R-LFP-180 (Fig. 2(c)). Overall, a LiOH/OA molar ratio of 2 : 1 was determined to be optimal for lithium replenishment in spent LFP at 150 $^{\circ}\text{C}$.

Considering the LiOH/OA molar ratio of 2 : 1 as a regeneration agent (RA), the ratio of S-LFP to RA is critical for practical applications. Doubling the amount of RA compared to R-LFP-150 resulted in R-LFP-150-3, where the XRD pattern (Fig. 2(f)) showed the presence of Li_3PO_4 impurities. These findings suggest that excessive RA promotes the formation of the Li_3PO_4 phase. To further optimize the process, we doubled the concentration of all reactants, leading to R-LFP-150-4. The XRD pattern (Fig. 2(g)) displayed distinct LFP-specific peaks, indicating that adjusting feedstock concentrations while maintaining the correct molar ratios is a promising strategy to reduce wastewater production during LFP direct regeneration.

We investigated the chemical valence changes of regenerated LFP (denoted as R-LFP) using X-ray photoelectron spectroscopy (XPS) (Fig. 3(a)). The results indicate that S-LFP exhibited Fe^{3+} peaks at approximately 712.00 eV and Fe^{2+} peaks at around 710.19 eV in the Fe 2p $_{3/2}$ region. After regeneration, R-LFP showed only Fe^{2+} peaks (710.35 eV), confirming that most Fe^{3+} was reduced during the hydrothermal process under the reductive influence of oxalic acid. The valence states of Fe were further verified by electron energy loss spectroscopy (EELS) (Fig. 4(b)). Changes in Fe valence states were evident from the energy loss shifts of the Fe L $_2$ -edge and L $_3$ -edge.³⁰ The lower shift of the Fe L $_3$ -edge in R-LFP-150 compared to S-LFP indicates the successful reduction of Fe^{3+} to Fe^{2+} during regeneration.

The morphology of R-LFP-150 and S-LFP was examined using scanning electron microscopy (SEM) and high-resolution transmission electron microscopy (HRTEM). The SEM images of R-LFP-150 (Fig. 3(c) and Fig. S3a, b (ESI †)) revealed a particle size comparable to that of S-LFP (Fig. 3(d) and Fig. S3c, d (ESI †)). The HRTEM image of R-LFP-150 (Fig. 3(e)) displayed well-defined lattice fringes with an interplanar spacing of 0.42 nm, corresponding to the (101) crystal plane of LFP.²⁰ The elemental composition of R-LFP-150 and S-LFP was analyzed *via* energy-dispersive X-ray spectroscopy (EDX) (Fig. S4, ESI †). Both materials exhibited signals for C, O, Fe, and P, while no detectable signals of fluorine (F) or aluminum (Al) were observed in S-LFP. This indicates that the regeneration process was highly effective, ensuring the complete removal of Al fragments, binder, and electrolyte salt from S-LFP.

To evaluate the electrochemical performance and regeneration efficacy of our method, we assembled 2032-type coin cells using R-LFP-120, R-LFP-150, R-LFP-180, and S-LFP as electrodes. As shown in Fig. 4(a), the R-LFP-150 exhibited an initial capacity of 163.1 mA h g $^{-1}$ at a 0.1C rate, which is significantly



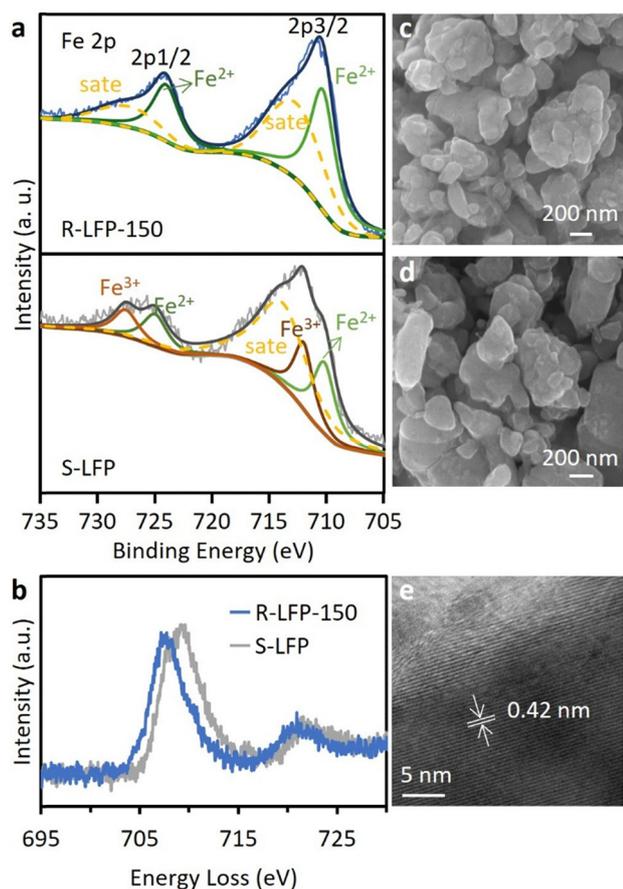


Fig. 3 (a) Fe 2p spectra of R-LFP-150 and S-LFP. (b) Fe L-edge EELS spectra of a representative R-LFP-150 and S-LFP. FESEM images of (c) R-LFP-150 and (d) S-LFP. (e) HRTEM of R-LFP-150.

higher than the $123.1 \text{ mA h g}^{-1}$ of S-LFP. Additionally, R-LFP-150 demonstrated a smaller potential interval compared to S-LFP (Fig. S5, ESI[†]) and a lower charge transfer resistance (R_{ct}) (Fig. S6, ESI[†]), highlighting the superior effect of the regeneration process.

The low-rate cycling performance was evaluated at a 0.2C rate (Fig. 4(b)). R-LFP-150 exhibited an initial capacity of $159.2 \text{ mA h g}^{-1}$ and maintained stable performance over 200 cycles, retaining $147.5 \text{ mA h g}^{-1}$ at the end. R-LFP-120 showed an initial capacity of $138.8 \text{ mA h g}^{-1}$, stabilizing at $130.4 \text{ mA h g}^{-1}$ after 200 cycles. Both R-LFP-150 and R-LFP-120 demonstrated significant capacity improvements compared to the S-LFP electrode, which delivered $116.8 \text{ mA h g}^{-1}$ in the first cycle and $103.5 \text{ mA h g}^{-1}$ after 200 cycles. The slightly lower capacity of R-LFP-120 compared to R-LFP-150 is attributed to the residual impurity phase of FePO_4 (as shown in Fig. 2(a)), which does not contribute to capacity.²⁷ In contrast, the R-LFP-180 electrode exhibited an initial capacity of $113.1 \text{ mA h g}^{-1}$, lower than that of S-LFP, which might be due to the formation of the new $\text{Fe}_5(\text{PO}_4)_4(\text{OH})_3 \cdot 2\text{H}_2\text{O}$ phase (Fig. 2(c)).

Fig. 4(c) and Fig. S7 (ESI[†]) illustrate the rate performance of R-LFP and S-LFP at various current rates (0.1, 0.2, 0.5, and 1C). The reversible specific capacities of R-LFP-150 were 163.1,

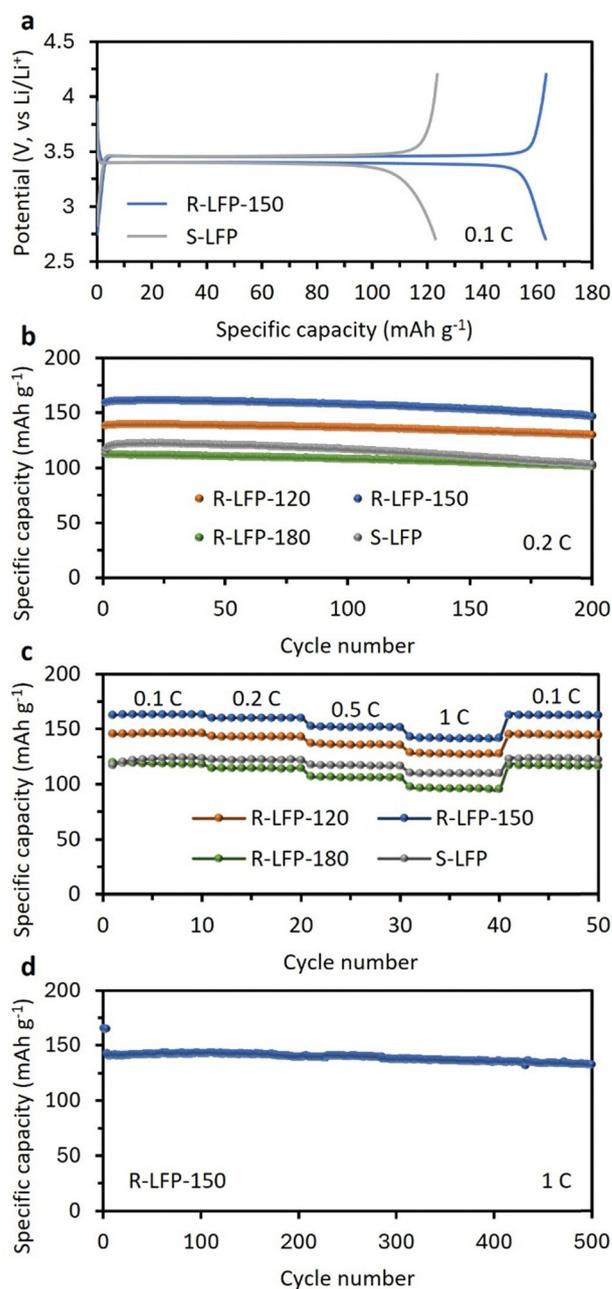


Fig. 4 (a) 0.1C Charge and discharge curves of R-LFP-150 and S-LFP. (b) 0.2C cycling performance and (c) rate performance comparison of R-LFP-120, R-LFP-150, R-LFP-180 and S-LFP. (d) long cycling performance of R-LFP-150.

160.2 , 151.6 , and $141.7 \text{ mA h g}^{-1}$, respectively. Upon returning to a 0.1C rate, a capacity of $162.5 \text{ mA h g}^{-1}$ was recovered, demonstrating superior rate performance compared to S-LFP, which exhibited specific capacities of 123.1 , 122.2 , 117.2 , and $109.7 \text{ mA h g}^{-1}$. Similarly, R-LFP-120 showed enhanced performance over S-LFP, delivering capacities of 146.0 , 143.1 , 135.9 , and $127.5 \text{ mA h g}^{-1}$. However, for R-LFP-180, the low-rate capacity was comparable to that of S-LFP, but at higher charging/discharging rates, the capacity declined more significantly, indicating that excessively high regeneration temperatures



adversely affect lithium replenishment and induce new phase generation, as demonstrated by the XRD pattern in Fig. 2(c).

We further investigated the high-rate (1C) cycling performance of R-LFP-150 (Fig. 4(d)). The electrode delivered 142.9 mA h g⁻¹ in the first 1C discharge. After 500 cycles, it retained a capacity of 133.2 mA h g⁻¹, with an exceptionally low degradation rate of 0.013%.

In summary, this study demonstrates the effectiveness of oxalic acid and lithium hydroxide as both reducing agents and lithium sources for the efficient regeneration of spent LiFePO₄ materials. The effects of reaction temperature, reagent dosage, structural composition, and electrochemical performance were systematically investigated and optimized. The regenerated LiFePO₄ (R-LFP-150) exhibited enhanced electrochemical performance, delivering an initial capacity of 159.2 mA h g⁻¹ at 0.2C and maintaining stability over 200 cycles. This method provides a viable approach for directly regenerating spent LiFePO₄ from various degradation states, significantly enhancing its feasibility for practical applications.

J. Z. Q. W. and Z. L. conceived the original concept and initiated the project. J. Z. mainly performed the cycling of cylindrical cells, chemical treatment, SEM, XRD, and electrochemical performance. S. Z. and N. T. helped partial experiments. H. T. carried out TEM and EELS test. W. W. carried out XPS test. J. Z. wrote the manuscript.

Data availability

All the data supporting this article have been included in the main text and the ESI.†

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Conflicts of interest

There are no conflicts to declare.

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